

21 August 1991

Lead Smelter and Refiner

Ohio Environmental Protection Agency
Division of Solid and Hazardous Waste Management
1800 WaterMark Drive
Columbus, Ohio 43266-1049

RE: Certification of Precompliance

Dear Sir:

On 21 May 1991 Master Metals, a secondary lead smelting facility, submitted a notification of possible hazardous waste activity based on the industrial furnace regulations.

As specified in Part 266.10 (c), operators of smelting, melting, and refining furnaces, that process hazardous waste solely for metal recovery are conditionally exempt from the industrial furnace regulations unless the hazardous waste has a total concentration of organic compounds listed in Part 261, Appendix VIII, exceeding 500ppm. One feedstock processed by Master Metals contains tetraethyl lead (TEL), an organic compound listed in Appendix VIII. The amount of TEL in the material varies and may at times exceed 500 ppm.

Master Metals had requested for advice as to whether Master Metals is exempt from Part 266.100 and if not exempt what additional standards are applicable to this situation. No response has been forthcoming from USEPA or OEPA on this matter.

Therefore, Master Metals is submitting an engineering study completed for Master Metals by a registered professional engineer. This study, using computer modeling and best engineering judgement, indicates that the smelting of this material meets the emission standards pertaining to organic emissions, particulate matter, specified metals, HCl, and Cl₂. In addition, Master Metals conducted stack compliance testing in March 1991, which indicated compliance with the lead, particulate, and sulfur dioxide emissions limitations.

-2-

Master Metals believes that the above is sufficient data to prove compliance with the industrial furnace emissions regulations, even though it is not at all clear whether these regulations pertain in this situation.

Please contact Mr. Rudy A. Zupan at (216) 621-2361 if you have any questions, comments, or require additional information.

Sincerely


Douglas K. Mickey
President

cc: Michael Cyphert, Esq.
Thompson, Hine, & Flory

USEPA, Region V
Waste Management Division 5HR-13

||||||| Hazen Research, Inc.
HAZEN 4601 Indiana St., Golden, CO 80403
||||||| 279-4501; FAX: 278-1528

#18853

FAX TRANSMITTAL

cc: Barry Hansen

Date: 7/16/90
Pages: 4

Time: 1330
Project: 6911

To: Mr. Douglas Mickey; Master Metals
Fax #: (216) 621-7475
From: Paul Queneau, Hazen Research
Re: Reply to your question, as posed by Chris Knopf (#3; 7/9/90)

Question: What is the interaction between sulfur dioxide emission levels and hazardous constituents in the slag?

Reply: Smelting lead-acid-battery wastes with soda ash and iron in short rotary furnaces typically permits capture of at least 98% the furnace charge's sulfur content as slag, matte, and dust. The balance of the sulfur reports to the gas phase entering the baghouse, generally as SO₂.

The silicate slag phase holds up to \approx 2% S, i.e., \approx 1/3 of the sulfur in your typical furnace charge. Up to \approx 1/2 of the sulfur in Master Metals' typical charge is captured as Na₂S·xFeS matte, in which is dissolved PbS and other metal sulfides, as well as FeO. For details on our understanding of the furnace metallurgy, refer to our publication, "Optimizing Matte and Slag Compositions In Rotary Furnace Smelting of Lead Residues."

Up to \approx 1/4 of the sulfur in a typical Master Metals' furnace charge first reports to the flue gas, then reacts with metal oxides in the gas, probably to form metal sulfates. The sulfates are collected in the baghouse as flue dust, e.g., as ZnSO₄ and PbSO₄, then subsequently recycled to the furnace.

Whether the slag-matte waste is hazardous via the EP or TCLP leach tests is hypothesized to be a function of the amount of Na₂O in the slag that is available to neutralize the acidic leachant. The solubility of lead is a minimum at about pH 11. The exact pH of this minimum depends on a number of solution variables, including temperature. Too much Na₂O results in increased lead solubility as Pb(OH)₃⁻; too little Na₂O results in increased lead solubility as Pb₆(OH)₈⁴⁺ and Pb₃(OH)₄²⁺. Refer to Figure 15.9 and Table 15.17, attached.



Illinois Environmental Protection Agency

2200 Churchill Road, Springfield, IL 62706

217/782-6760

Refer to: 0312580003 -- Cook County
Riverdale/Acme Steel Co.
Waste Stream Permit #830796

May 19, 1988

David J. Holmberg
Manager - Environmental Services
Acme Steel Company
13500 S. Perry Avenue
Riverdale, Illinois 60627-1182

Dear Mr. Holmberg:

In response to your letter of April 26, 1988 and our phone conversations, please be informed that the charcoal dross currently permitted for disposal in permit 830796 can be considered an "industrial material" if it is used as a feedstock for ASARCO or any other similar lead user and not a solid waste. This material can be transferred for users or re-users without being subject to manifest or annual report requirements, and the material is not subject to 90 day limitation on storage without a permit. This in no way relieves Acme Steel from properly containing, labeling, storing and transporting this industrial material in a proper manner and in accordance with all other pertinent rules.

If you have any further questions, please give me a call. If Acme Steel produces any other industrial materials that may be of use for other industries, or if Acme Steel can use other industries' samples or "waste" materials, consider additional uses of the Industrial Material Exchange Service (IMES, or the "waste exchange") for help with these transactions. I've enclosed a copy of the latest IMES publication.

Sincerely,


Michael F. Nechvatal, Manager
Solid Waste Management Section
Division of Land Pollution Control

MFN:bjh/145BJ/113

Enclosure

cc: Division File
L. Eastep, Permits
G. Savage, FOS
H. Chappel, Compliance
J. Morgan, IMES

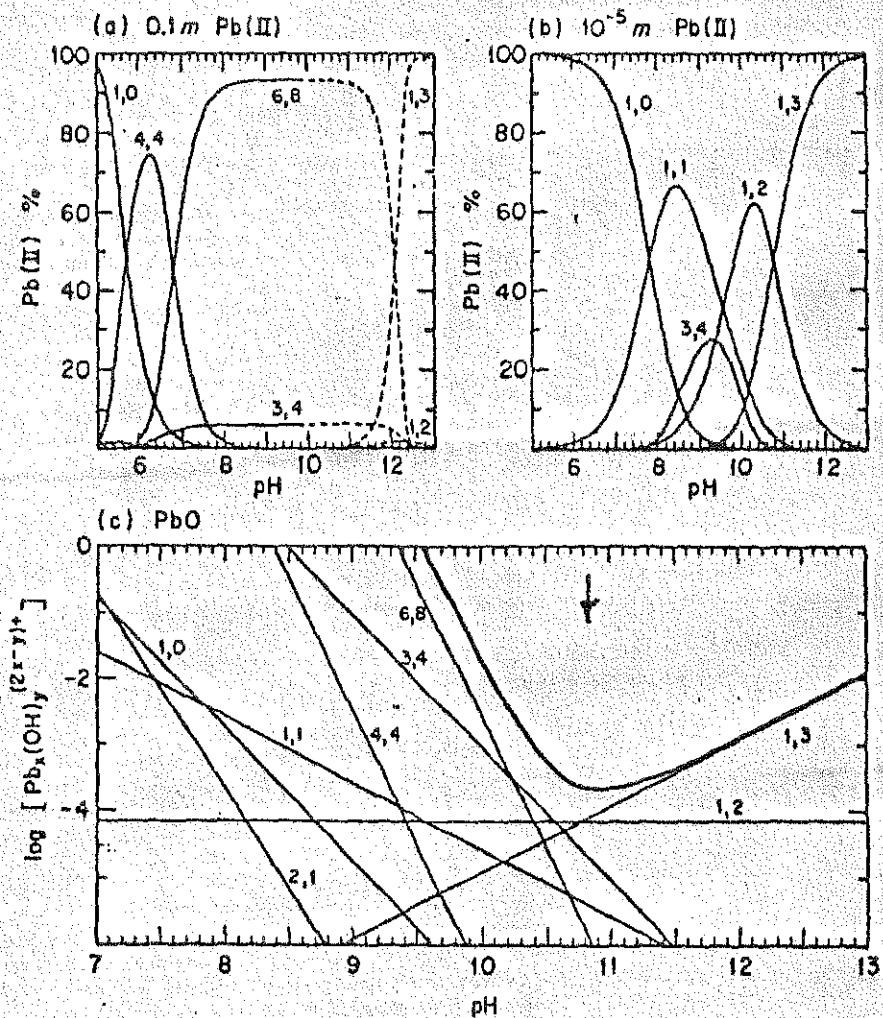


Fig. 15.9. Distribution of hydrolysis products (x, y) at $I = 1 \text{ m}$ and 25° in (a) 0.1 m $\text{Pb}(\text{II})$, (b) 10^{-5} m $\text{Pb}(\text{II})$, and (c) solutions saturated with PbO . The dashed curves in a and b denote regions supersaturated with respect to PbO ; the heavy curve in c is the total concentration of lead(II).

15.4.5 Summary. The hydrolysis behavior of Pb^{2+} ion has been the subject of extensive work involving a large number of physical chemical techniques. This ion is a model case in which the species identification from most of the information is quite consistent. There are seven species (including the uncomplexed Pb^{2+}) which exist under widely varying conditions (see Table 15.17); their distribution is shown in Fig. 15.9. At least four of the hydrolysis species may be considered to be definitely identified, and this is possible because of the

Table 15.17. Summary of Pb^{2+} hydrolysis at $25^\circ C$

$$\log Q_{xy} = \log K_{xy} + a^{1/2}/(1 + j^{1/2}) + bm_x$$

Species or phase	$\log K_{xy}$	a	b			$\alpha(\log Q_{xy})$
			$m_x = 0.3$	$m_x = 1.0$	$m_x = 3.5$	
$PbOH^+$	-7.71	-1.022	0.93	0.47	0.156	± 0.1
$Pb(OH)_2$	-17.12	-1.022	1.0	0.4	0.09	± 0.1
$Pb(OH)_3^-$	-28.06	0	0.67	0.0	-0.21	± 0.05
Pb_2OH^{3+}	-6.36	1.022	-0.5	-0.33	-0.17	± 0.1
$Pb_3(OH)_4^{2+}$	-23.88	-2.044	4.17	2.13	0.71	± 0.2
$Pb_4(OH)_4^{4+}$	-20.88	2.044	0.85	0.41	0.10	± 0.1
$Pb_6(OH)_8^{4+}$	-43.61	0	3.18	1.59	0.48	± 0.1
$PbO(c)$ (red) ($\log Q_{xy}$)	12.72	1.022	0.08	-0.05	-0.05	± 0.05

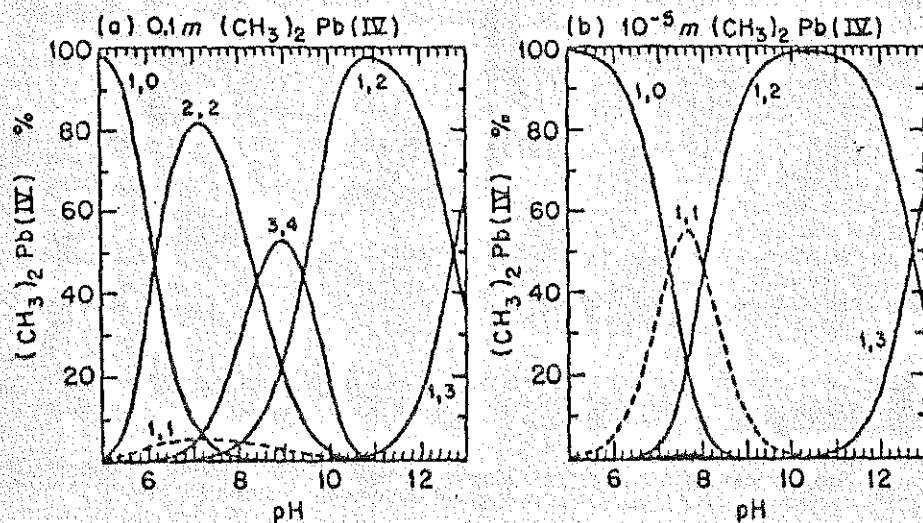


Fig. 15.10. Distribution of dimethyllead(IV) and species (x,y) at $I = 1\text{ m}$ and $25^\circ C$ in (a) $0.1\text{ m } (\text{CH}_3)_2\text{Pb(IV)}$ and (b) $10^{-5}\text{ m } (\text{CH}_3)_2\text{Pb(IV)}$.

wide range of compositions which are available for study. The most prominent polynuclear species are $Pb_4(OH)_4^{4+}$ and $Pb_6(OH)_8^{4+}$. Hydroxyl numbers up to about 1.33 and concentrations of $Pb(II)$ in excess of 2 M are attainable. The stable oxide phase at $25^\circ C$ is red tetragonal PbO . Hydrolysis of $(\text{CH}_3)_2\text{Pb}^{2+}$ occurs above pH of 5, and the distribution of species in 1 m salt medium is shown in Fig. 15.10. Pb^{4+} hydrolyzes very extensively, but its hydrolysis products are not well known. $Pb(OH)_6^{2-}$ is probably the anionic species occurring in base.

The Hydrolysis of Cations

CHARLES F. BAES, Jr.

ROBERT E. MESMER

Oak Ridge National Laboratory
Oak Ridge, Tennessee

1976

A WILEY-INTERSCIENCE PUBLICATION

JOHN WILEY & SONS, New York • London • Sydney • Toronto

HAZEN **Hazen Research, Inc.**
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279-4501; FAX: 278-1528

*Revised
Dwy*
*** 18667**

FAX TRANSMITTAL

cc: Tom Helms

Date: 7/5/90
Pages: 8

Time: 1030
Project: 6911

To: Mr. Douglas Mickey; Master Metals
Fax #: (216) 621-7475

From: Paul Queneau, Hazen Research
Re: Your question on adding soda ash to your main baghouse
to limit SO₂ emissions

Adding soda ash (Na₂CO₃) to the gas stream entering your primary baghouse appears to be a winner. That small portion of the sulfur that escapes fixation in the slag can thus be captured. Furthermore, the soda ash provides additional alkalinity to neutralize H₂SO₄ that may form due to excursions below the dew point. For a general description of dry scrubbing practice, in this case using nahcolite (NaHCO₃, i.e., sodium bicarbonate), refer to the attached paper by N.D. Shah.

At Carteret, we added lime to the blast furnace flue for SO₂ capture; it worked well. What makes your application using soda ash particularly attractive is that you have 100% dust recycle back to the furnaces. Sodium values added to the baghouse are thus fully utilized in the furnace.

Note that dry scrubbing with soda ash has fundamental differences from SO₂ collection using nahcolite. Nahcolite operates best above 300°F to facilitate its decomposition to soda ash. This decomposition process generates low density Na₂CO₃ having high active surface area. However, when using soda ash directly, the Na₂CO₃ must already be in an active form to provide efficient SO₂ collection.

The following are suggestions that you may wish to consider.

- Key for SO₂ collection is using crude light soda ash, which has high surface area. Collection efficiency using your dense soda ash will almost

Re: Na_2CO_3 to BH to SO_2 capture



$$\text{Stoic. } \text{Na}_2\text{CO}_3/\text{SO}_2 = 1.67/1$$

$$\text{" } \text{Na}_2\text{SO}_3/\text{Na}_2\text{CO}_3 = 1.19/1$$

From 6/14% mass & heat balance,
estimated 3.0 lb SO_2 /hr from small
furnace. Assume both furnaces are
operating, and that combination of
 Na_2CO_3 hold-up in baghouse, plus use
of two furnaces, maintains average
ratio of $\text{Na}_2\text{CO}_3/\text{SO}_2$ in baghouse.

$$(3.0 + 1.2(3.0)) = 6.6 \text{ lb } \text{SO}_2/\text{hr, max.}$$

Now assume that you assign 10% of current
solids through ret in BH to Na_2CO_3 , and
 Na_2SO_3 (50/50 ratio), May also form some Na_2S_4 .

Current dust load $\approx 415 \text{ lb dust/hr, i.e.,}$

$$\left(\frac{4.5 \text{ scoops dust}}{\text{full charge}} \right) \left(\frac{550 \text{ lb dust}}{\text{scoop}} \right) \left(\frac{4 \text{ charge}}{\text{day}} \right) \left(\frac{1 \text{ day}}{24 \text{ hrs}} \right) = 413.8/\text{hr}$$

Therefore can increase solids in BH by
415 lb/hr, or add $(41.5)((1.0+1.19)/2)$

$$= 37.9 \text{ lb } \text{Na}_2\text{CO}_3 \text{ added to BH/hr,}\\ \text{say } 40 \text{ lb/hr}$$

from 6/14/90 mass & heat balance,
& from 5/17/90 charge calculation,
average Na_2CO_3 fed to off furnace
 ≈ 5 scoops / total heat, new furnace
takes "6" scoops.

$$\left(\frac{5.5 \text{ scoop } \text{Na}_2\text{CO}_3}{\text{Ave. total heat}} \right) \left(\frac{700 \text{ lb } \text{Na}_2\text{CO}_3}{\text{scoop}} \right) \left(\frac{4 \text{ heat}}{\text{day}} \right) \left(\frac{1 \text{ day}}{24 \text{ hrs}} \right) = 642 \text{ lb } \text{Na}_2\text{CO}_3 \text{ used/hr.}$$

Therefore we are to be using $(37.9/642)(100)$,
or about 6% of plant Na_2CO_3 to BH,
which is reasonable.

If stoichiometric RX of Si_2 with Na_2CO_3 ,
can capture $(37.9)/(1.67) = 22.7 \text{ lb } \text{SiO}_2/\text{hr.}$

Only expect 6.6 lb $\text{SiO}_2/\text{hr.}$, max.

Therefore use of ~40 lb $\text{Na}_2\text{CO}_3/\text{hr}$
to BH seems reasonable for starters.

Soda ash is recovered during dust
cycle. Expect to fix 298% of recycled
 Na_2CO_3 in stay-mate.

Plant analyses should show whether more,
or less Na_2CO_3 , should be added to BH,
once you have operated with the 6 lb $\text{Na}_2\text{CO}_3/\text{hr.}$

(2)

1. D. McDay
1. May 1982

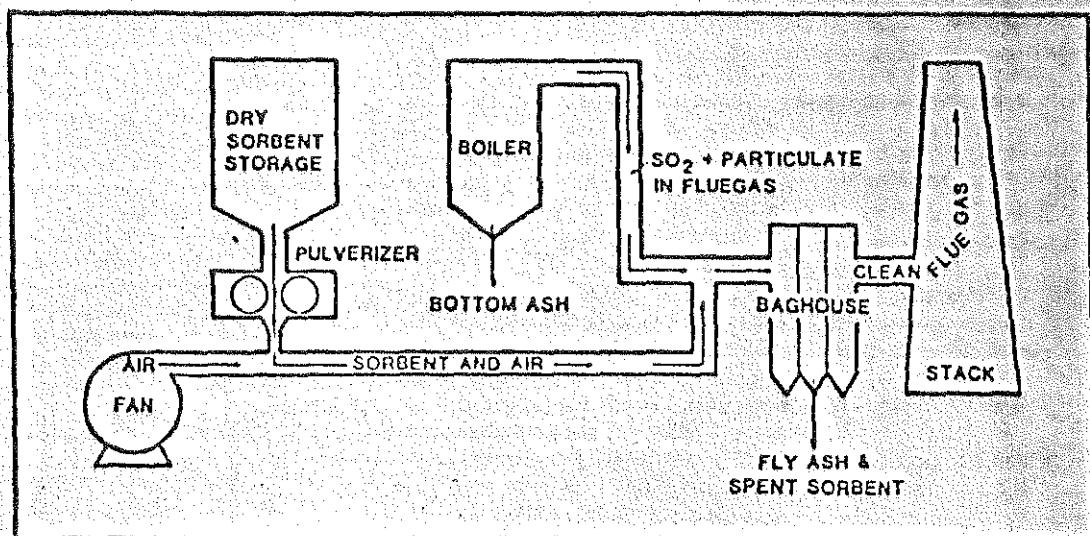


Figure 1. Dry-scrubbing process.

POLLUTION CONTROL PRACTICES:

Dry Scrubbing of SO_2

Among the advantages of dry over wet scrubbing are reduced corrosion erosion, and scaling, and capital costs that are 30 to 50% lower.

N. D. Shah,

Multi Mineral Corp., Grand Junction, Co. 81501

It is advantageous to use dry scrubbing processes to remove SO_2 from flue gases. There are three major flue gas desulfurization processes: wet scrubbing, spray drying, and dry scrubbing.

The factors influencing the selection of a dry-scrubbing process are analyzed in this article. Nahcolite is used as the dry-scrubbing agent for this analysis, because nahcolite (a naturally occurring sodium bicarbonate mineral) is shown to be the most likely candidate for future commercial applications. However, most of the conclusions about nahcolite scrubbing are valid for processes employing other dry-scrubbing agents, if allowances are made for their reduced reactivity, increased cost, etc.

Advantages of dry scrubbing

Wet scrubbing of SO_2 , using an alkaline solution is a

Table 1. Factors influencing scrubber selection.

1. Sulfur Content of the Coal
2. Type of Particulate Control
3. Water Requirements
4. Location of the Plant
5. SO_3 Removal
6. NO_x Removal
7. Energy Consumption
8. Cost
9. Operating Experience

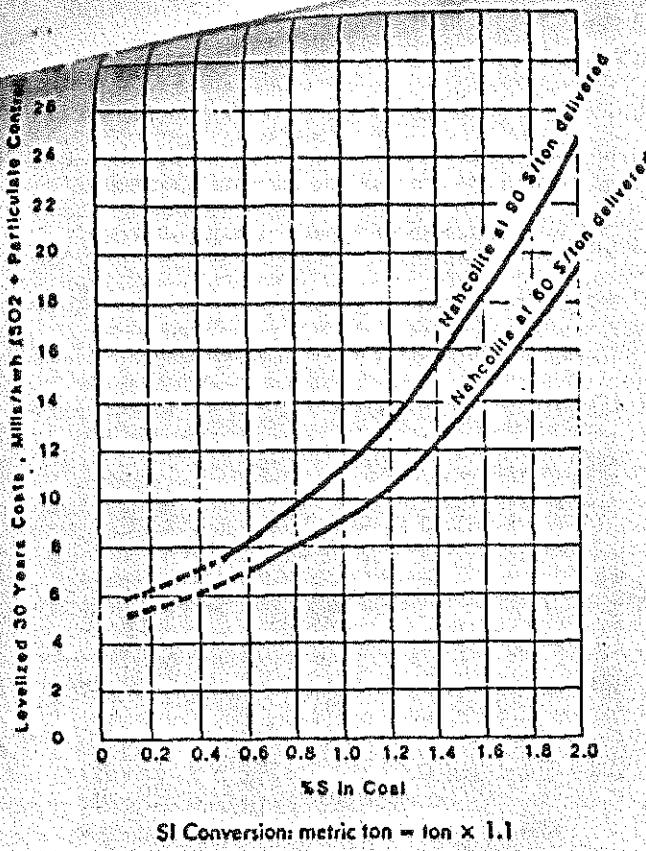


Figure 2. Cost of dry scrubbing vs. %S in coal.

well-known technology. A wealth of data are available in the literature concerning the commercial operation of wet scrubbing systems. Even though wet scrubber technology has been commercially proven, process reliability has often been disappointing. Corrosion, erosion, plugging, scaling and freezing have been identified as major problems with wet-scrubbing system.

Because the majority of the wet scrubber problems relate to the use of circulating alkaline water solutions, spray dryer technology has been introduced to reduce water requirements by about one half. In the spray-drying process, a solution is sprayed into the flue gas stream. The sprayed material reacts with SO_2 and at the same time water evaporates (from the heat of the flue gas) resulting in dry waste. The dried material also reacts with the SO_2 during collection in the particulate control device.

Even though spray dryer technology appears more promising than wet scrubbing, a number of questions concerning the commercial reliability of the process remain unanswered. These questions relate to reheat requirements and plugging, scaling and corrosion problems resulting from the use of water. Approximately 1,600 MW of new coal-fired generating capacity under construction has committed to spray drying for SO_2 control. There is no doubt that spray dryer technology offers the potential of reducing or minimizing wet scrubber problems.

Dry-scrubbing processes carry spray-drying philosophy one step further by eliminating water usage entirely. In a dry-scrubbing process, a dry alkaline powder is injected, Figure 1, directly into the flue gas stream. The alkaline particles react with SO_2 while suspended in the gas stream. The waste product from this reaction is subsequently collected in a particulate collection device (baghouse, precipitator, etc.) and the clean flue gas is vented to the atmosphere.

Additional reaction between alkaline particles captured in the particulate control equipment and SO_2 occurs as the flue gas passes through the particulate control equipment.

Although the reactivity of dry scrubbing agents will naturally be somewhat lower than the reactivity of wet-sorbent solutions used in wet scrubbing or spray drying, a dry-scrubbing processes offers four clear advantages:

1. The absence of water results in the minimization or elimination of corrosion, erosion, plugging, scaling and freezing problems.

2. Dry-scrubbing processes control SO_2 and particulates in a single piece of equipment. The capital costs of the process are projected to be 30 to 50% lower than wet scrubber or spray dryer systems. Operating costs are projected to be approximately equivalent to spray dryers.

3. Utilities are familiar with dry material handling systems (for coal and flyash); therefore, a dry-scrubbing process will not pose the problem of adopting an unknown technology.

4. Wet scrubbers require flue gas reheating and result in high pressure drops across the system. The dry-scrubbing process eliminates both the reheating requirements and high-pressure drop conditions, thus resulting in 3 to 5% energy savings based on plant energy.

Disposal techniques

Three areas of concern presently exist for dry-scrubbing processes: the commercial availability of scrubbing agents, the lack of commercial operating data, and the problem of waste disposal.

Dry-scrubbing agents that have been tested in the past are trona, soda ash, commercial sodium bicarbonate, lime, limestone and nahcolite (sodium bicarbonate in mineral form). Based on past dry-scrubbing tests (1-3), the best scrubbing agents are commercial sodium bicarbonate, trona, and nahcolite. Economic constraints dictate that commercial sodium bicarbonate is too expensive at its present cost (4) of about \$230/ton (\$254/m ton).

Trona is commercially mined for the production of soda ash. Soda ash manufacturers are reluctant to sell raw trona because tax depletion allowances for trona result in unattractive investment return. At present, nahcolite appears to be the most promising scrubbing agent for dry scrubbing. Nahcolite is not mined commercially today, but Multi Mineral Corp. is presently working to develop a commercial operation to produce one million ton/yr (907,000 m ton/yr) of nahcolite by early 1986.

The operating experience with the dry-scrubbing process has been extended to demonstration tests at the 22-MW scale using nahcolite as the scrubbing agent. These tests are currently being conducted by Public Service Co. of Colorado, Electric Power Research Institute and Multi Mineral Corp. at the Cameo coal-fired generating station. The Cameo test results (5) have been very encouraging, and have demonstrated the consistent attainment of SO_2 removals of over 70% at nahcolite stoichiometric ratio of less than one.

The principle constituents of the waste produced from the nahcolite dry-scrubbing process are sodium bicarbonate, carbonate and sulfate. These compounds are relatively water-soluble and may require runoff protection. Many disposal techniques have been proposed for sodium waste disposal, including the following:

- Ocean dumping
- Deep well disposal
- Insolubilization by aqueous coprecipitation
- Insolubilization by sintering
- Land disposal (dry arid basins)
- Use of the salt cake as a byproduct
- Landfill in clay isolation cells

A detailed discussion of these techniques has been pre-

The most cost-effective means of disposal is by land disposal or landfill in clay cells.

Factors influencing selection

The selection of an SO₂ scrubbing system depends on technical, economic and regulatory factors some of which are shown in Table 1.

Sulfur Content of the Coal. Sulfur content of the coal determines the injection rate of the scrubbing agent, which influences the design size of the particulate collection equipment, the material handling and injection system, and the waste disposal facility. All these factors impact the economics of the process. Figure 2 shows the economic sensitivity (7) of the dry (nahcolite) scrubbing process to sulfur content in the coal. It appears that the 30-year levelized cost of a dry process increases rapidly with the increase in sulfur content of the coal.

The SO₂ removal requirements as defined by the revised (1979) new source performance standard is based on the sulfur level in the coal. Figure 3 shows the SO₂ removal requirements as a function of % sulfur in the coal with a heating value of 10,000 Btu/lb (23,300 kJ/kg).

Figure 4 shows the SO₂ removal efficiency (7) of the dry-scrubbing process as a function of nahcolite stoichiometric ratio. The increased costs of the dry-scrubbing process at higher sulfur contents in the coal partly arises from the increased nahcolite requirements to achieve greater SO₂ removal efficiencies. Consequently, the dry-scrubbing process is most promising for lower sulfur coals.

Type of Particulate Control Equipment. The electrostatic precipitator, baghouse or wet scrubber are the three most common particulate control equipment used in a coal-fired plant. Since the purpose of the dry-scrubbing process is to keep the flue gas dry, using a wet scrubber for particulate control is undesirable. The major disadvantage with the electrostatic precipitator is that it does not provide the additional contact time between SO₂ and the dry sorbent during the collection of the sorbent. If a baghouse is the preferred method of particulate control, the dry-scrubbing process should definitely be considered as a means of SO₂ control.

Wet scrubbers, although effective in removing SO₂ and/or particulates, often create secondary problems related to visible emissions and violation of opacity regulations. The dry injection process with a baghouse obviates these problems by keeping the flue gas in the dry state.

Water Requirements. Wet scrubbers and spray dryers consume water at a rate of approximately 1 gpm/MW (3.8 L/min · MW) and $\frac{1}{2}$ gpm/MW (1.9 L/min · MW), respectively. The dry injection process does not use any water and is, therefore, a top candidate for plants in areas where water is a scarce commodity (western U.S.).

Location of the Plant. The application of the dry-scrubbing process for SO₂ control may be limited to the western part of the country for the following reasons:

- The only known deposit of nahcolite, which is the most promising dry-scrubbing agent, is found in Western Colorado. As a result, transportation costs to users in the west will be less.

- A dry-scrubbing system eliminates water usage; since water is a precious commodity in the west, it is logical to consider application of this technology for the western U.S.

- The use of baghouses for particulate control is becoming more popular in the western U.S., because the resistivity of the low-sulfur coal ash is too high for effective collection by precipitators. A dry-scrubbing process with a baghouse offers combined SO₂ and particulate control in one piece of equipment.

- Waste disposal problems are less acute due to the semi-arid climate.

SO₂ Removal. Past studies (8) have shown that the dry-scrubbing process will remove SO₂ in the same proportion

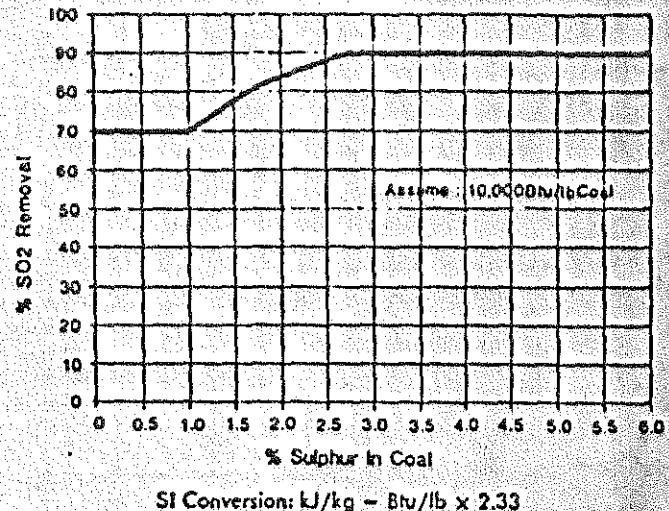


Figure 3. EPA regulations for SO₂ control.

as SO₃. Wet scrubbers do not remove SO₃, but create problems related to corrosion, erosion, etc., when SO₃ comes in contact with water. The dry-scrubbing process should be considered in cases where high SO₃ concentrations are in the flue gas.

NO_x Removal. Pilot-scale tests (2,5) have shown that dry scrubbing with nahcolite removes approximately 10 to 40% of the NO_x contained in the flue gas stream. Although further testing needs to be conducted to confirm this result, there is a possibility of simultaneous SO₂ plus NO_x removal with dry scrubbing. This would allow SO₂/NO_x/particulate control in one equipment.

Energy Consumption. Wet scrubbers consume about 3 to 5% of a plant's energy in reheating the flue gas and overcoming the pressure drop across the system. The dry-scrubbing process robs very little of the plant's energy, resulting in substantial savings.

Cost. Figure 5 shows the capital cost for SO₂ and particulate control among the wet scrubber, spray dryer and dry-

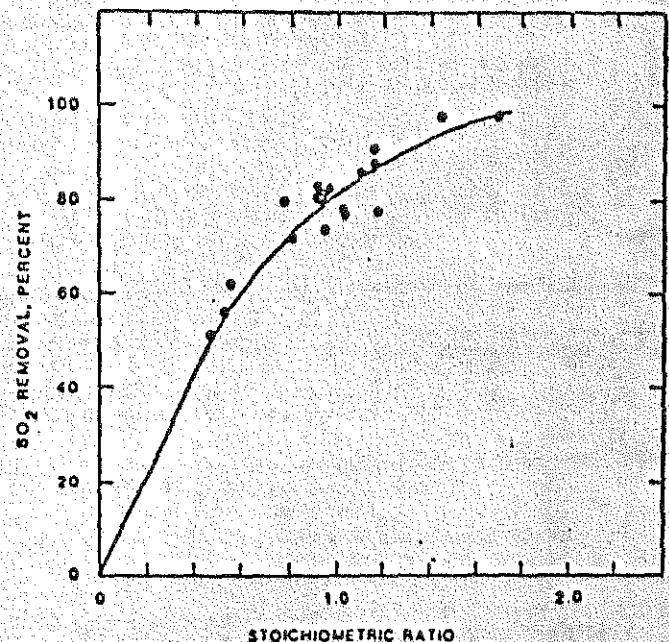


Figure 4. SO₂ removal efficiency.

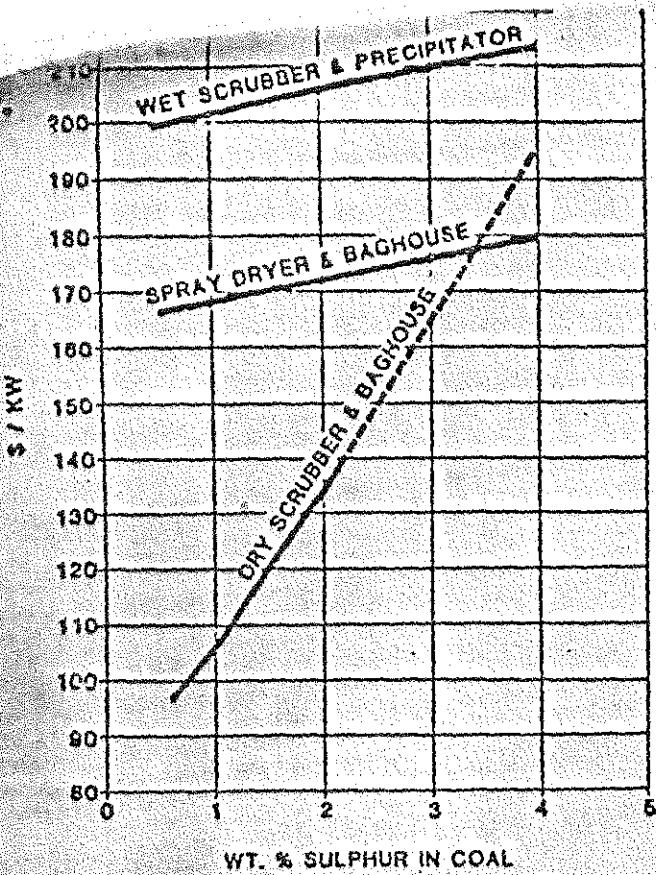


Figure 5. Capital cost comparison.

scrubbing process as a function of percentage sulfur in coal. Based on these results, the dry-scrubbing process is attractive for coals containing less than 3% sulfur.

The leveled costs of an SO₂ control system depend largely on the sulfur content of the coal and reagent costs. The costs of the wet scrubber, spray dryer and dry-scrubbing process as a function of these variables are shown in Figure 6. The dry-scrubbing process appears attractive for lower-sulfur, coal-fired installations. At a nahcolite cost of \$90/ton (\$99/m ton) delivered, the dry-scrubbing process appears attractive for plants burning less than 0.9% sulfur coal.

Operating Experience. The dry-scrubbing process has a limited operating history. Most of its operating data is based on pilot- and bench-scale testing. To date, the process has been successfully demonstrated at the 22-MW scale. But the simplicity of the dry-scrubbing process should significantly impact the amount of operating experience required to establish the technology as a commercial reality. The spray dryer, also a relatively new technology, has little commercial operating data available. Wet scrubbers have been in commercial use for more than a decade.

Commercial nahcolite production

Multi Mineral Corp. is presently pursuing commercial production of nahcolite from its resource interest. An initial production of nahcolite at one million ton per year (909,000 ton/yr) by early 1986 is planned. This production level will be sufficient to scrub 70% of the SO₂ from approximately 5,000 MW of generating capacity burning 1% sulfur oil.

An upgrading process, which results in a nahcolite purity approximately 80%, has been tested on a pilot scale. We plan to market a product which will be approximately 80% sodium bicarbonate and minus 1/4 in. (0.6 cm) in size. Other

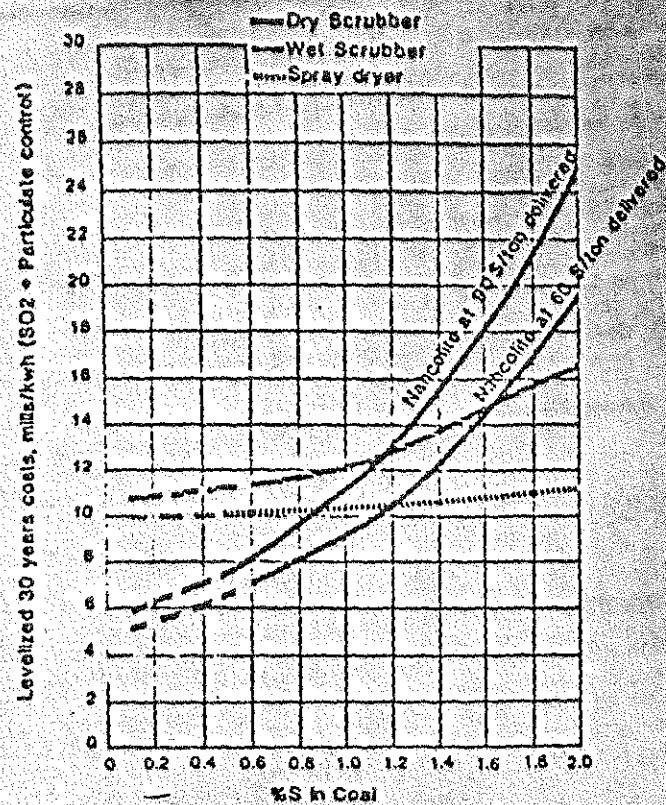


Figure 6. Cost comparison.

sizes and grades may become available as the commercial operation evolves.

It is estimated that the price of the nahcolite will vary depending on the product specifications, but will probably range from \$50 to \$100/ton (\$55 to 110/m ton) for a nahcolite product containing approximately 80% sodium bicarbonate.

Literature cited

1. "Evaluation of Dry Alkalies for Removing Sulfur Dioxide from Boiler Flue Gases," EPRI Report No. FP-207 (Oct., 1976).
2. Han Liu, et al., "Evaluation of Fabric Filter as Chemical Contractor for Control of Sulfur Dioxide from Flue Gas," NTIS Report No. PB 194 196 (Dec., 1969).
3. Shah, N.D., D.P. Teixeira, and R.C. Carr, "Application of Dry Sorbent Injection for SO₂ and Particulate Control," Paper Presented at EPA FGD Symp. (Nov., 1977).
4. Chem. Marketing Reporter, 220, No. 12, p. 46 (Sept., 1981).
5. Muzio, L.J., et al., "Demonstration of SO₂ Removal on a 22-MW Coal-Fired Utility Boiler by Dry Injection of Nahcolite," Paper Presented at EPRI Conf. on Fabric Filter Tech. (July, 1981).
6. "Evaluation of Dry Sorbent and Fabric Filtration for FGD," EPA Report No. 68-02-1412 by TRW (Jan., 1977).
7. Lapp, D.E., et al., "Use of Nahcolite for Coal-Fired Power Plants," Paper Presented at 7th National Conf. on Energy and Environment (Dec., 1980).
8. "Bench-Scale Study of the Dry Removal of SO₂ with Nahcolite and Trona," EPRI Report No. CS-1744 (March, 1981).



N. D. Shah is manager of Nahcolite use development with the Multi Mineral Corp. A holder of a B.S.Ch.E. degree from the Univ. of Bombay, a M.S.Ch.E. from the Univ. of North Dakota, and a Ph.D. degree in chemical engineering from the Colorado School of Mines, he has been involved with pilot testing and evaluation of dry scrubbing technology for eight years. Prior to joining the firm, he worked as a project manager of dry scrubbing activities for Electric Power Research Institute.

certainly be very low. You should be able to purchase the light soda ash from Kerr-McGee.

Kerr-McGee generates light ash ($\approx 98\% \text{Na}_2\text{CO}_3$) in its bicarbonate calciners at Searls Lake (Trona, CA). Light ash is a process intermediate, rather than an item of commerce. I understand, however, that it may be obtainable for about a 20% price premium over your current soda ash cost.

- Sufficient flue-gas velocity ~~and~~ is needed to prevent the fine soda ash added to the flue from settling out in the flue before it reaches the baghouse. The longer the soda ash has to contact the flue gas, the greater should be the efficiency of SO_2 capture.
- At Carteret, we fed fine lime to the flue through a square flange installed on the top of the round flue. Upon the flange was mounted a hopper equipped with vibrator. This approach may also be appropriate for injecting soda ash. A slide gate controlled flow of the fine solids into the flue, as assisted by the negative pressure in the flue.
- Most of the SO_2 fixation is likely to occur while the gas flows through the soda ash mixed with the flue dust on the bags. The SO_2 collection rate improves somewhat with temperature, indicating that the soda ash might best be added to the diluted gas just before the convective cooler ($\approx 640^\circ\text{F}$). At this temperature, no fusion problem is anticipated. Mechanical constraints in your plant may necessitate that the soda ash be dispersed in the flue after the cooler immediately upstream from the baghouse.
- Calculations (attached) indicate that you could begin at a Na_2CO_3 injection rate of $\approx 40 \text{ lb Na}_2\text{CO}_3/\text{hr}$. Please check the various tonnages that went into the calculations to be sure that they in fact represent your current soda ash (640 lb/hr), maximum SO_2 (6.6 lb/hr), and dust (415 lb/hr) tonnage. I assumed that 2% of the sulfur fed to the furnace reported to the flue as SO_2 , and that there was $\approx 10\%$ extra capacity in the bags to accept the Na_2CO_3 - Na_2SO_3 - Na_2SO_4 solids.

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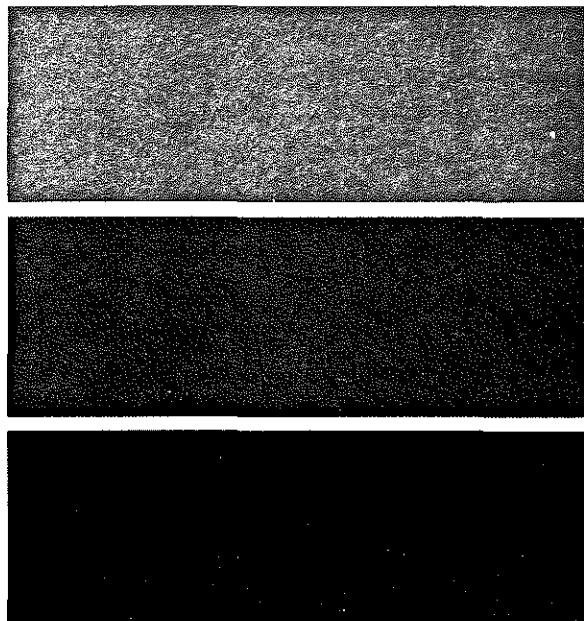
P.O. Box 152, Richfield, Ohio 44286
Phone (216) 526-0990

REPORT NO. 91-1174 1603
COMPANY Master Metals
TITLE Compliance
DATE 3-12-91

**MASTER METALS INC.
CLEVELAND, OHIO**

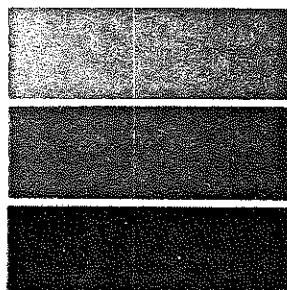
**FURNACE & FUGITIVE BAGHOUSES
PARTICULATE, SULFUR DIOXIDE & LEAD
EMISSION EVALUATION**

CONDUCTED - MARCH 12, 1991



SOURCE EVALUATION RESULTS

PREPARED BY



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Phone (216) 526-0990

March 27, 1991

Mr. Douglas K. Mickey
President
Master Metals Incorporated
2850 West Third Street
Cleveland, Ohio 44113

Dear Mr. Mickey:

The following report is the result of the particulate, sulfur dioxide, and lead emission evaluation conducted on March 12, 1991 at the above location. Three (3) test runs were conducted on this date at the exhaust stacks of the two (2) baghouses serving the lead smelter. The systems are designated Furnace Baghouse and Fugitive Baghouse.

The results are true and accurate to the degree specified in the pertinent sections of the Code of Federal Regulations, in force at the time of testing.

I am looking forward to answering any questions you may have and assisting you in the future.

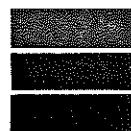
Respectfully submitted,



Tom E. Holder
Environmental Engineer
ENVISAGE ENVIRONMENTAL INC.

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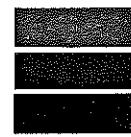
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INTRODUCTION



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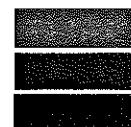
INTRODUCTION

On March 12, 1991, Envisage Environmental Inc. conducted an emission evaluation at Master Metals Incorporated, 2850 West Third Street, Cleveland, Ohio. Testing was conducted at the exhaust stacks of the two (2) baghouses used for air pollution control of the lead smelting operation. Test parameters included particulate, sulfur dioxide, and lead conducted at each location.

The purpose of these tests was to determine compliance with applicable State and Federal Regulations concerning air pollution emissions. The furnace and the baghouses were monitored by Master Metals Inc. and EPA personnel. Test parameters were in accordance with USEPA Reference Methods 1-6 and 12.

The Envisage testing team consisted of Messrs. Bob Hovan, Joe Dossa, Mark Gierke, and Steve Norris. The Ohio EPA was represented by Mr. Douglas Seaman, Division of Air Pollution Control, Cleveland, Ohio. Mr. Rudy Zupan, Master Metals Inc., coordinated the testing.

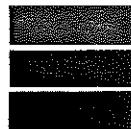
Results are presented in this report for particulate, sulfur dioxide, and lead emissions with the various velocity, volumetric and temperature measurements associated with these tests for each baghouse.



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DESCRIPTION OF PROGRAM



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DESCRIPTION OF PROGRAM

The evaluation consisted of three (3) test runs, each one (1) hour in length. At each location, six (6) sample points were used in each of the two (2) ports. Sample time per point was five (5) minutes for a total sample time per run of sixty (60) minutes. Diagrams of the sample point locations are included in this report.

The samples were withdrawn from the gas streams isokinetically through three (3) foot Pyrex lined probes. The entire lengths of the probes were heated and attached to a Method 5 sample train modified for the collection of sulfur dioxide and lead. The hot boxes were set to maintain a temperature of 248 degree F. and the heated areas were monitored to ensure condensation did not form prior to the impingers. Exit gas temperature of the impingers was maintained below 68 degrees F. with an ice bath. The nozzle, probe and connecting glassware were cleaned before testing and at the conclusion of each test run with acetone. Leak checks of the pitot tube lines and the sample trains were all acceptable by EPA regulations. Cyclonic flow was less than ten (10) degrees at each location.

The method 5 impinger train was modified by replacing the distilled water with the following: # 1 - 100 ml 80% Isopropanol, # 2 - 200 ml 3% Hydrogen Peroxide, # 3 - 100 ml 0.1 Nitric Acid, # 4 - empty, # 5 - 200 grams silica gel. Analysis for sulfur dioxide was by the barium-thorin titration method. The filters and impinger solutions were analyzed for lead by atomic absorption spectrometry.

Flue gas analysis was conducted by drawing an integrated air bag sample throughout each test run and analyzed with a Hays Republic Model 621A "Orsat" Portable Gas Analyzer. The average of at least three readings for each run were used in calculating the emission rates.



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Description of Program - continued

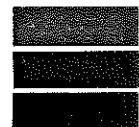
Calibration of the equipment used, including the dry gas meters, orifice meters, and the "S" type pitot tubes were conducted within 60 days of the test date. Copies of the data are included in this report.

All analytical procedures were performed in accordance with the methods specified in the Code of Federal Regulations, Title 40 Part 60, Volume 43. Blanks were collected and analyzed on the distilled water and acetone used in the evaluation. The residue from the distilled water was less than could be measured on a 0.1 milligram analytical balance and was considered zero. The acetone blank was recorded and incorporated into the results.

The example equations included in this report represents the data collected during Run # 1 conducted at the Fugitive Baghouse exhaust.

TEST RESULTS

SUMMARY



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TEST RESULTS SUMMARY

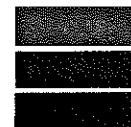
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Master Metals Inc
2850 West 3rd Street
Cleveland, Ohio

Furnace Baghouse - Particulate,
Lead, & Sulfur Dioxide Emission Evaluation

Conducted - March 12, 1991

PARAMETER	RUN # 1	RUN # 2	RUN # 3
Particulate Emissions			
Pounds/hour	0.25	0.41	0.24
Grains/dscf	0.0021	0.0033	0.0019
Lead Emissions - Pounds/hour	0.009	0.004	0.002
Sulfur Dioxide Emissions			
Pounds/hour	0.06	* BDL	* BDL
Pounds/dscf	6.88E-08	* BDL	* BDL
ppmV	0.4	* BDL	* BDL
System Flow Rates			
Feet/second	80.61	80.71	79.27
ACFM	15,835	15,854	15,572
DSCFM	14,082	14,474	14,649
Moisture Content			
Volume percent	2.73	1.66	1.24
Sample Location Temperature			
Degrees Fahrenheit	112	104	89
* Below Detectable Limit			



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TEST RESULTS SUMMARY

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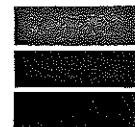
Master Metals Inc
 2850 West 3rd Street
 Cleveland, Ohio

Fugitive Baghouse - Particulate,
 Lead, & Sulfur Dioxide Emission Evaluation

Conducted - March 12, 1991

PARAMETER	RUN # 1	RUN # 2	RUN # 3
Particulate Emissions			
Pounds/hour	0.82	0.78	0.50
Grains/dscf	0.0024	0.0023	0.0015
Lead Emissions - Pounds/hour	0.015	0.013	0.014
Sulfur Dioxide Emissions			
Pounds/hour	0.09	* BDL	* BDL
Pounds/dscf	3.71E-08	* BDL	* BDL
ppmV	0.2	* BDL	* BDL
System Flow Rates			
Feet/second	91.81	91.85	92.04
ACFM	38,937	38,953	39,034
DSCFM	39,989	39,887	40,053
Moisture Content			
Volume percent	1.31	1.07	1.04
Sample Location Temperature			
Degrees Fahrenheit	54	57	56

* Below Detectable Limit

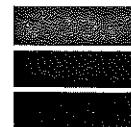


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TEST RESULTS

DETAILED



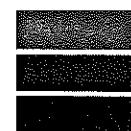
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TEST RESULTS
 Master Metals Inc
 Furnace Baghouse - Particulate,
 Lead, & Sulfur Dioxide Emission Evaluation

DATE:	March 12, 1991	Symbol	Units	RUN # 1	RUN # 2	RUN # 3
	Time of Day			0934 1046	1121 1226	1253 1357
1	Gas Volume-dry, std.	Vmstd	cu. ft.	47.65	48.11	49.21
2	Condensate Vapor Vol.	Vwstd	cu. ft.	1.34	0.81	0.62
3	Gas Stream Moisture	Bws	vol.dec	0.0273	0.0166	0.0124
4	Mol.Wt-flue gas (dry)	Msd	lb/lb mo.	28.84	28.84	28.84
5	Mol.Wt-flue gas (wet)	Ms	lb/lb mo.	28.54	28.66	28.70
6	Flue Gas Velocity	Vs	ft/sec	80.61	80.71	79.27
7	Flue Gas Volume-Actual	ACFM	cu. ft.	15,835	15,854	15,572
8	Flue Gas Volume-Std.	DSCFM	cu. ft.	14,082	14,474	14,649
9	Particulate Conc.	Cs				
	- Probe		gr/dscf	0.0019	0.0031	0.0018
	- Filter		gr/dscf	0.0002	0.0002	0.0001
	- Lead		gr/dscf	7.4E-05	3.4E-05	1.7E-05
	- Total		gr/dscf	0.0021	0.0033	0.0019
10	Emission Rate	E				
	- Probe		lb/hr	0.23	0.39	0.22
	- Filter		lb/hr	0.02	0.02	0.02
	- Lead		lb/hr	0.009	0.004	0.002
11	- Total		lb/hr	0.25	0.41	0.24
	Isokinetic Rate	I	%	96.3	94.6	95.6

* Totals DO NOT Include Lead



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TEST RESULTS

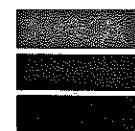
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Master Metals Inc

Fugitive Baghouse - Particulate,
Lead, & Sulfur Dioxide Emission Evaluation

DATE:	March 12, 1991	Symbol	Units	RUN # 1	RUN # 2	RUN # 3
	Time of Day			0940 1045	1120 1224	1252 1356
1	Gas Volume-dry, std.	Vmstd	cu. ft.	59.71	58.51	59.51
2	Condensate Vapor Vol.	Vwstd	cu. ft.	0.79	0.64	0.63
3	Gas Stream Moisture	Bws	vol.dec	0.0131	0.0107	0.0104
4	Mol.Wt-flue gas (dry)	Msd	lb/lb mo.	28.84	28.84	28.84
5	Mol.Wt-flue gas (wet)	Ms	lb/lb mo.	28.69	28.72	28.72
6	Flue Gas Velocity	Vs	ft/sec	91.81	91.85	92.04
7	Flue Gas Volume-Actual	ACFM	cu. ft.	38,937	38,953	39,034
8	Flue Gas Volume-Std.	DSCFM	cu. ft.	39,989	39,887	40,053
9	Particulate Conc.	Cs				
	- Probe		gr/dscf	0.0021	0.0022	0.0013
	- Filter		gr/dscf	0.0003	0.0001	0.0001
	- Lead		gr/dscf	4.4E-05	3.9E-05	4.1E-05
	- Total		gr/dscf	0.0024	0.0023	0.0015
10	Emission Rate	E				
	- Probe		lb/hr	0.74	0.74	0.45
	- Filter		lb/hr	0.09	0.04	0.04
	- Lead		lb/hr	0.015	0.013	0.014
11	- Total		lb/hr	0.82	0.78	0.50
	Isokinetic Rate	I	%	91.7	90.1	91.3

* Totals DO NOT Include Lead



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SO2 LABORATORY SUMMARY
 Master Metals Inc
 Furnace Baghouse - Particulate,
 Lead, & Sulfur Dioxide Emission Evaluation

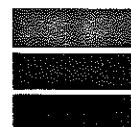
DATE: March 12, 1991	Symbol	Units	RUN # 1	RUN # 2	RUN # 3
20 Normality of Ba(ClO ₄) ₂	N	meq/ml	0.0097	0.0097	0.0097
21 Volume of solution	Vsln	ml	240.0	240.0	230.0
22 Volume aliquot titrant	Va	ml	20.00	20.00	20.00
23 Volume Ba(ClO ₄) ₂ Blank	Vtb	ml	0.0	0.0	0.0
24 Volume Ba(ClO ₄) ₂ Sampl	Vt	ml	0.40	0.00	0.00

SO2 TEST RESULTS

Master Metals Inc
 Furnace Baghouse - Particulate,
 Lead, & Sulfur Dioxide Emission Evaluation

DATE: March 12, 1991		Units	RUN # 1	RUN # 2	RUN # 3
12 Concentration SO ₂	C _{SO2}	lb/dscf	6.88E-08	* BDL	* BDL
13 Concentration SO ₂	PPM	ppmV	0.4	* BDL	* BDL
14 Emission Rate SO ₂	E _{SO2}	lb/hr	0.06	* BDL	* BDL

* Below Detectable Limit



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SO2 LABORATORY SUMMARY
 Master Metals Inc
 Fugitive Baghouse - Particulate,
 Lead, & Sulfur Dioxide Emission Evaluation

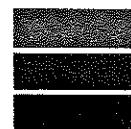
DATE: March 12, 1991	Symbol	Units	RUN # 1	RUN # 2	RUN # 3
20 Normality of Ba(ClO ₄) ₂	N	meq/ml	0.0097	0.0097	0.0097
21 Volume of solution	V _{sln}	ml	240.0	230.0	225.0
22 Volume aliquot titrant	V _a	ml	20.00	20.00	20.00
23 Volume Ba(ClO ₄) ₂ Blank	V _{tb}	ml	0.0	0.0	0.0
24 Volume Ba(ClO ₄) ₂ Samp1	V _t	ml	0.27	0.00	0.00

SO2 TEST RESULTS

Master Metals Inc
 Fugitive Baghouse - Particulate,
 Lead, & Sulfur Dioxide Emission Evaluation

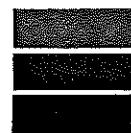
DATE: March 12, 1991	Units	RUN # 1	RUN # 2	RUN # 3
12 Concentration SO ₂	C _{SO2}	lb/dscf	3.71E-08	* BDL
13 Concentration SO ₂	PPM	ppmV	0.2	* BDL
14 Emission Rate SO ₂	E _{SO2}	lb/hr	0.09	* BDL

* Below Detectable Limit



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SAMPLE POINT LOCATION DIAGRAM



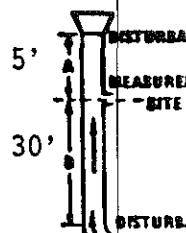
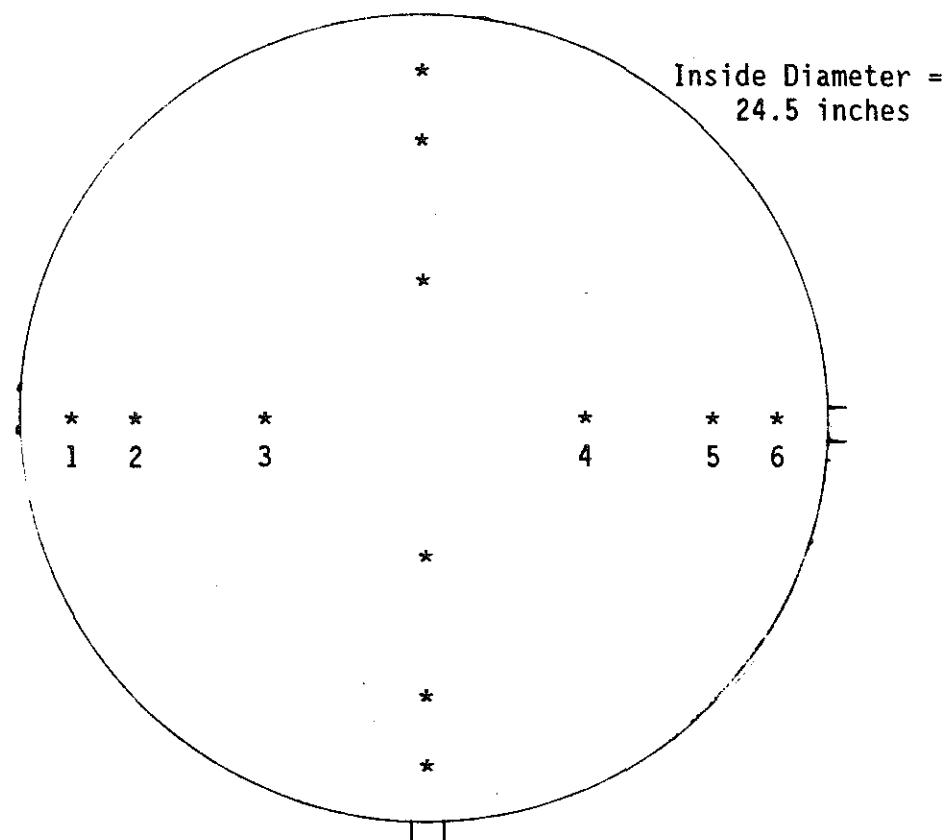
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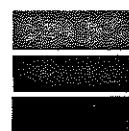
SAMPLE POINT LOCATIONS

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Master Metals
Furnace Baghouse Exhaust



<u>Point Number</u>	<u>Distance from Inside Wall</u>
1	23.4 inches
2	20.9 inches
3	17.2 inches
4	7.3 inches
5	3.6 inches
6	1.1 inches



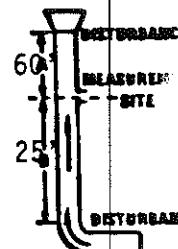
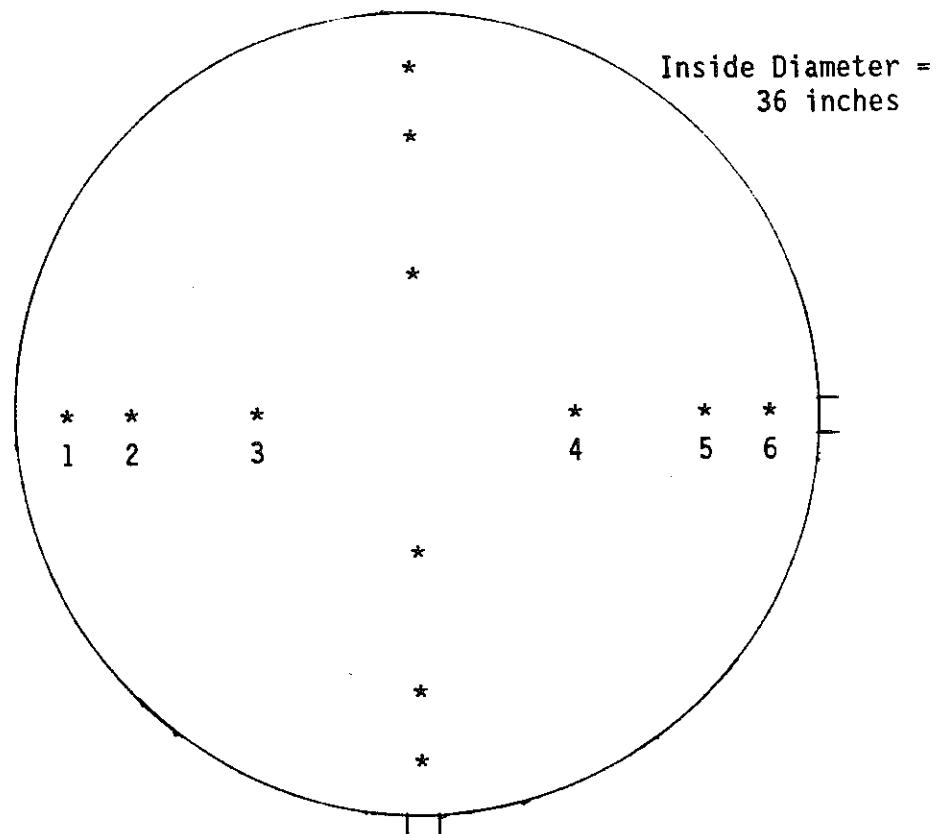
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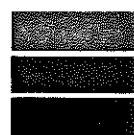
SAMPLE POINT LOCATIONS

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Master Metals
Fugitive Baghouse Exhaust



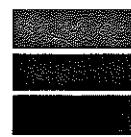
<u>Point Number</u>	<u>Distance from Inside Wall</u>
1	34.4 inches
2	30.7 inches
3	25.3 inches
4	10.7 inches
5	5.3 inches
6	1.6 inches



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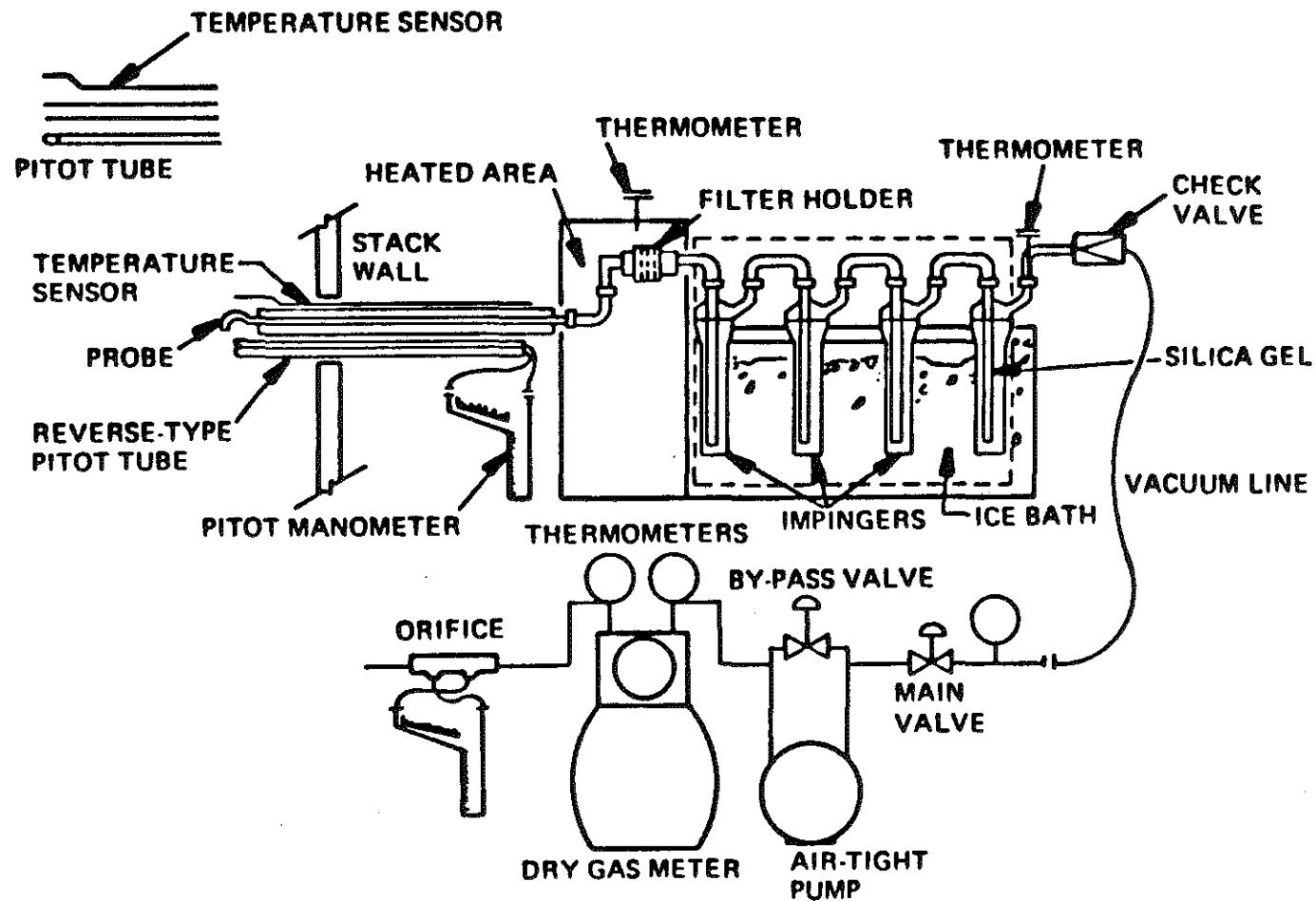
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SAMPLING TRAIN DIAGRAM



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EPA METHOD 5 PARTICULATE SAMPLE APPARATUS

LABORATORY SECTION

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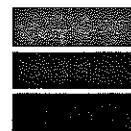
LABORATORY SUMMARY SHEET

Master Metals Inc

Furnace Baghouse - Particulate,

Lead, & Sulfur Dioxide Emission Evaluation

DATE: March 12, 1991	Symbol	Units	RUN # 1	RUN # 2	RUN # 3
1 Sampling Time	t	minutes	60.0	60.0	60.0
2 Barometric Pressure	Pb	in. Hg	30.18	30.18	30.18
3 Static Pressure	Pg	in. H2O	-7.20	-7.20	-7.20
Stack Pressure	Ps	in. Hg	29.65	29.65	29.65
4 Gas Meter Volume	Vm	cu. ft.	50.30	52.53	53.44
5 Stack Area	A	sq. ft.	3.27	3.27	3.27
6 Nozzle Diameter	Dn	dec. in.	0.1875	0.1875	0.1875
7 Meter Temperature	Tm	degrees F	105.0	124.5	121.4
8 Stack Temperature	Ts	degrees R	565.0	584.5	581.4
	Ts	degrees F	112.3	103.6	89.3
9 Velocity Head	[^] P	in. H2O	1.365	1.380	1.374
10 Orifice Pressure	[^] H	in. H2O	2.11	2.08	2.15
11 Carbon dioxide	CO2	%	0.0	0.0	0.0
12 Oxygen	O2	%	20.9	20.9	20.9
13 Carbon monoxide	CO	%	0.0	0.0	0.0
14 Nitrogen	N2	%	79.1	79.1	79.1
15 Pitot Coefficient	Cp		0.84	0.84	0.84
16 Water Collected	Vlc	ml	28.4	17.3	13.1
Sample Weight:	Mn				
17 - Probe		g	0.0059	0.0098	0.0057
18 - Filter		g	0.0006	0.0005	0.0004
19 - Lead		mg	0.22974	0.10752	0.05289



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Phone (216) 526-0990

LABORATORY SUMMARY SHEET

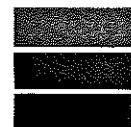
22

Master Metals Inc

Fugitive Baghouse - Particulate,

Lead, & Sulfur Dioxide Emission Evaluation

DATE: March 12, 1991	Symbol	Units	RUN # 1	RUN # 2	RUN # 3
1 Sampling Time	t	minutes	60.0	60.0	60.0
2 Barometric Pressure	Pb	in. Hg	30.18	30.18	30.18
3 Static Pressure	Pg	in. H2O	1.90	1.90	1.90
Stack Pressure	Ps	in. Hg	30.32	30.32	30.32
4 Gas Meter Volume	Vm	cu. ft.	59.95	59.45	59.95
5 Stack Area	A	sq. ft.	7.07	7.07	7.07
6 Nozzle Diameter	Dn	dec. in.	0.1875	0.1875	0.1875
7 Meter Temperature		degrees F	78.8	85.3	80.7
	Tm	degrees R	538.8	545.3	540.7
8 Stack Temperature		degrees F	54.2	56.9	56.0
	Ts	degrees R	514.2	516.9	516.0
9 Velocity Head	[^] P	in. H2O	1.663	1.660	1.665
10 Orifice Pressure	[^] H	in. H2O	3.10	3.13	3.14
11 Carbon dioxide	CO2	%	0.0	0.0	0.0
12 Oxygen	O2	%	20.9	20.9	20.9
13 Carbon monoxide	CO	%	0.0	0.0	0.0
14 Nitrogen	N2	%	79.1	79.1	79.1
15 Pitot Coefficient	Cp		0.84	0.84	0.84
16 Water Collected	Vlc	ml	16.8	13.5	13.3
Sample Weight:	Mn				
17 - Probe		g	0.0083	0.0082	0.0051
18 - Filter		g	0.0010	0.0004	0.0005
19 - Lead		mg	0.17072	0.14720	0.15735



**Envisage
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Incorporated**

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Phone (216) 526-0990

PLANT Master Metals, Furnace Baghouse

DATE March 12, 1991

25

RUN NO. 1

CASE NO. 3

CONTAINER NUMBER	WEIGHT OF PARTICULATE COLLECTED		
	FINAL WEIGHT	TARE WEIGHT	WEIGHT GAIN
352	0.6168	0.6162	0.0006
N/A	- -	- -	- -
202	105.3060	105.3001	0.0059

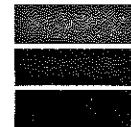
* Corrected for Acetone Blank

VOLUME OF LIQUID WATER COLLECTED		
	IMPIINGER VOLUME (ml)	SILICA GEL WEIGHT (g)
FINAL	418	241.6
INITIAL	400	231.2
NET LIQUID COLLECTED	18	10.4
TOTAL NET VOLUME	28.4	* g ml

*

Convert weight of water to volume by dividing weight increase by density of water:

$$\frac{\text{Increase g}}{(1 \text{ g/ml})} = \text{Volume Water, ml}$$



**Envisage
Environmental
Incorporated**

P.O. Box 152 Richfield, Ohio 44286
Phone (216) 526-0990

PLANT Master Metals, Furnace Baghouse
 DATE March 12, 1991
 RUN NO. 2
 CASE NO. 7

24

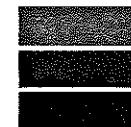
CONTAINER NUMBER	WEIGHT OF PARTICULATE COLLECTED		
	FINAL WEIGHT	TARE WEIGHT	WEIGHT GAIN
246	0.5765	0.5760	0.0005
N/A	--	--	--
302	106.1179	106.1081	0.0098

* Corrected for Acetone Blank

VOLUME OF LIQUID WATER COLLECTED		
	IMPIINGER VOLUME (ml)	SILICA GEL WEIGHT (g)
FINAL	404	244.5
INITIAL	400	231.2
NET LIQUID COLLECTED	4	13.3
TOTAL NET VOLUME	17.3	*
		g ml

* Convert weight of water to volume by dividing weight increase by density of water:

$$\frac{\text{Increase g}}{(1 \text{ g/ml})} = \text{Volume Water, ml}$$



**Envisage
Environmental
Incorporated**
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PLANT Master Metals, Furnace Baghouse
 DATE March 12, 1991
 RUN NO. 3
 CASE NO. 77

25

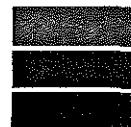
CONTAINER NUMBER	WEIGHT OF PARTICULATE COLLECTED		
	FINAL WEIGHT	TARE WEIGHT	WEIGHT GAIN
268	0.5900	0.5896	0.0004
N/A	- -	- -	- -
1	162.9029	162.8972	0.0057

* Corrected for Acetone Blank

VOLUME OF LIQUID WATER COLLECTED		
	IMPIINGER VOLUME (ml)	SILICA GEL WEIGHT (g)
FINAL	405	239.3
INITIAL	400	231.2
NET LIQUID COLLECTED	5	8.1
TOTAL NET VOLUME	13.1	* g ml

* Convert weight of water to volume by dividing weight increase by density of water:

$$\frac{\text{Increase g}}{(1 \text{ g/ml})} = \text{Volume Water, ml}$$



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Phone (216) 526-0990

PLANT Master Metals, Fugitive Baghouse

DATE March 12, 1991

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RUN NO. 1

CASE NO. 8

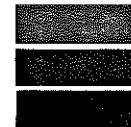
CONTAINER NUMBER	WEIGHT OF PARTICULATE COLLECTED		
	FINAL WEIGHT	TARE WEIGHT	WEIGHT GAIN
270	0.5905	0.5895	0.0010
N/A	- -	- -	- -
301	108.9172	108.9089	0.0083

* Corrected for Acetone Blank

VOLUME OF LIQUID WATER COLLECTED		
	IMPIINGER VOLUME (ml)	SILICA GEL WEIGHT (g)
FINAL	402.6	246.3
INITIAL	400	232.1
NET LIQUID COLLECTED	2.6	14.2
TOTAL NET VOLUME	16.8	* g ml

* Convert weight of water to volume by dividing weight increase by density of water:

$$\frac{\text{Increase g}}{(1 \text{ g/ml})} = \text{Volume Water, ml}$$



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PLANT Master Metals, Fugitive Baghouse
 DATE March 12, 1991
 RUN NO. 2
 CASE NO. 14

27

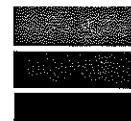
CONTAINER NUMBER	WEIGHT OF PARTICULATE COLLECTED		
	FINAL WEIGHT	TARE WEIGHT	WEIGHT GAIN
245	0.5794	0.5790	0.0004
N/A	- -	- -	- -
168	108.0401	108.0319	0.0082

* Corrected for Acetone Blank

VOLUME OF LIQUID WATER COLLECTED		
	IMPIINGER VOLUME (ml)	SILICA GEL WEIGHT (g)
FINAL	400	245.6
INITIAL	400	232.1
NET LIQUID COLLECTED	0	13.5
TOTAL NET VOLUME	13.5	* g ml

* Convert weight of water to volume by dividing weight increase by density of water:

$$\frac{\text{Increase g}}{(1 \text{ g/ml})} = \text{Volume Water, ml}$$



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PLANT Master Metals, Fugitive Baghouse
 DATE March 12, 1991
 RUN NO. 3
 CASE NO. 20

28

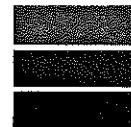
CONTAINER NUMBER	WEIGHT OF PARTICULATE COLLECTED		
	FINAL WEIGHT	TARE WEIGHT	WEIGHT GAIN
247	0.5805	0.5800	0.0005
N/A	- -	- -	- -
603	106.6528	106.6477	0.0051

* Corrected for Acetone Blank

VOLUME OF LIQUID WATER COLLECTED		
	IMPIINGER VOLUME (ml)	SILICA GEL WEIGHT (g)
FINAL	400	245.4
INITIAL	400	232.1
NET LIQUID COLLECTED	0	13.3
TOTAL NET VOLUME	13.3	* g ml

* Convert weight of water to volume by dividing weight increase by density of water:

$$\frac{\text{Increase g}}{(1 \text{ g/ml})} = \text{Volume Water, ml}$$



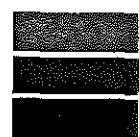
**Envisage
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Incorporated**

P.O. Box 152 Richfield, Ohio 44266
 Phone (216) 526-0990

ORSAT ANALYSIS DATA

Company Master Metals Incorporated
 Location Furnace Baghouse Exhaust
 Date March 12, 1991

	RUN # 1	RUN # 2	RUN # 3
ORSAT DATA			
Sample 1			
CO2 Reading	0.0	0.0	0.0
O2 Reading	20.9	20.9	20.9
CO Reading	20.9	20.9	20.9
Sample 2			
CO2 Reading	0.0	0.0	0.0
O2 Reading	20.9	20.9	20.9
CO Reading	20.9	20.9	20.9
Sample 3			
CO2 Reading	0.0	0.0	0.0
O2 Reading	20.9	20.9	20.9
CO Reading	20.9	20.9	20.9
AVERAGES			
% CO2	0.0	0.0	0.0
% O2	20.9	20.9	20.9
% CO	0	0	0
% N2	79.1	79.1	79.1

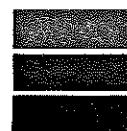


**Envisage
Environmental
Incorporated**
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 Phone (216) 528-0990

ORSAT ANALYSIS DATA

Company Master Metals Incorporated
 Location Fugitive Baghouse Exhaust
 Date March 12, 1991

	RUN # 1	RUN # 2	RUN # 3
ORSAT DATA			
Sample 1			
CO2 Reading	0.0	0.0	0.0
O2 Reading	20.9	20.9	20.9
CO Reading	20.9	20.9	20.9
-----	-----	-----	-----
Sample 2			
CO2 Reading	0.0	0.0	0.0
O2 Reading	20.9	20.9	20.9
CO Reading	20.9	20.9	20.9
-----	-----	-----	-----
Sample 3			
CO2 Reading	0.0	0.0	0.0
O2 Reading	20.9	20.9	20.9
CO Reading	20.9	20.9	20.9
-----	-----	-----	-----
AVERAGES			
% CO2	0.0	0.0	0.0
% O2	20.9	20.9	20.9
% CO	0	0	0
% N2	79.1	79.1	79.1



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KUDUKIS WASTEWATER LABS, INC.

2779 BROADWAY AVENUE, CLEVELAND, OHIO 44115

(216) 696 - 0280 FAX: (216) 696 - 6831

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ANALYSIS CERTIFIED BY: John Ondo

JO Laboratory Manager

Client ENVISAGE ENVIRONMENTAL
P.O. BOX 152
RICHFIELD, OHIO 44286
ATTN: AUDREY

Report Date: March 18, 1991
P.O. #: _____
Client: _____
Sample Received: 3/14/91
Lab Sample #: 1-0762-0764

Sample I.D.: #1 - EEI - Filter 246
#2 - EEI - Filter 268
#3 - EEI - Filter 270

ANALYSIS	#1	#2	#3	#4	ANALYSIS	#1	#2	#3	#4
Metals:	Total mg/Filter	Total mg/Filter	Total mg/Filter	—	Acidity (CaCO ₃)				
Cadmium					Alkalinity (CaCO ₃)				
Chromium: Total					Bacteria:				
Hexavalent					F. Coliform/100ml				
Trivalent					T. Coliform/100ml				
Copper					Total Plate Count/1ml				
Lead	0.06	0.03	0.14		Chloride				
Nickel					Chlorine: T. Res.				
Silver					Conductivity - uMHOS/cm				
Zinc					Cyanide: Total				
Aluminum					Amenable				
Antimony					Free				
Arsenic					Reactive				
Barium					Flash Point °F				
Beryllium					Fluoride				
Calcium					Hardness (CaCO ₃)				
Iron					MBAS				
Magnesium					Nitrogen: Nitrate (N)				
Manganese					Nitrite (N)				
Mercury					Ammonia (N)				
Potassium					T Kjeldahl (N)				
Selenium					Oil & Grease				
Silicon					Oxygen Demand: BOD ₅				
Sodium					COD				
Thallium					pH				
Tin					Phenols				
					Phosphorus: Total (P)				
					Residue: Total				
					Total Volatile				
					Suspended				
					Volatile Suspended				
					Sulfate				
					Sulfide: Total				
					Reactive				

ug/L = micrograms/Liter(ppb) • mg/L = milligrams/Liter(ppm) • mg/kg = milligrams/kilogram • > = greater than • < = less than (below detection)

Sampled By: Client xx KWLI Auto Sampler Other

KUDUKIS WASTEWATER LABS, INC.

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32

ANALYSIS CERTIFIED BY: John Ondo

JO Laboratory Manager

Client

ENVISAGE ENVIRONMENTAL
 P.O. BOX 152
 RICHFIELD, OHIO 44286

ATTN: AUDREY

Report Date: March 20, 1991

P.O. #: _____

Client: _____

Sample Received: 3/14/91

Lab Sample #: 1-0750-0752

Sample I.D.: #1 - EEI - OR1-1, 3/12/91
 #2 - EEI - OR1-2, 3/12/91
 #3 - EEI - OR1-3, 3/12/91

ANALYSIS	#1	#2	#3	#4	ANALYSIS	#1	#2	#3	#4
Metals:	Total mg/L	Total mg/L	Total mg/L	—	Acidity (CaCO ₃)				
Cadmium					Alkalinity (CaCO ₃)				
Chromium: Total					Bacteria:				
Hexavalent					F. Coliform/100ml				
Trivalent					T. Coliform/100ml				
Copper					Total Plate Count/1ml				
Lead	0.58	<0.05	0.16		Chloride				
Nickel					Chlorine: T. Res.				
Silver					Conductivity - uMHOS/cm				
Zinc					Cyanide: Total				
Aluminum					Amenable				
Antimony					Free				
Arsenic					Reactive				
Barium					Flash Point °F				
Beryllium					Fluoride				
Calcium					Hardness (CaCO ₃)				
Iron					MBAS				
Magnesium					Nitrogen: Nitrate (N)				
Manganese					Nitrite (N)				
Mercury					Ammonia (N)				
Potassium					T Kjeldahl (N)				
Selenium					Oil & Grease				
Silicon					Oxygen Demand: BOD ₅				
Sodium					COD				
Thallium					pH				
Tin					Phenols				
					Phosphorus: Total (P)				
					Residue: Total				
					Total Volatile				
					Suspended				
					Volatile Suspended				
					Sulfate				
					Sulfide: Total				
					Reactive				

ug/L = micrograms/Liter(ppb) • mg/L = milligrams/Liter(ppm) • mg/kg = milligrams/kilogram • > = greater than • < = less than (below detection)

Sampled By: Client XX KWLI _____ Auto Sampler _____ Other _____

KUDUKIS WASTEWATER LABS, INC.

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ANALYSIS CERTIFIED BY: John Ondo

JO Laboratory Manager

Client

ENVISAGE ENVIRONMENTAL
 P.O. BOX 152
 RICHFIELD, OHIO 44286

ATTN: AUDREY

Report Date: March 20, 1991

P.O. #: _____

Client: _____

Sample Received: 3/14/91

Lab Sample #: 1-0753-0755

Sample I.D.: #1 - EEI - OR2-1, 3/12/91
 #2 - EEI - OR2-2, 3/12/91
 #3 - EEI - OR2-3, 3/12/91

ANALYSIS	#1	#2	#3	#4	ANALYSIS	#1	#2	#3	#4
Metals:	Total mg/L	Total mg/L	Total mg/L	—	Acidity (CaCO ₃)				
Cadmium					Alkalinity (CaCO ₃)				
Chromium: Total					Bacteria:				
Hexavalent					F. Coliform/100ml				
Trivalent					T. Coliform/100ml				
Copper					Total Plate Count/1ml				
Lead	1.24	<0.05	<0.05		Chloride				
Nickel					Chlorine: T. Res.				
Silver					Conductivity - uMHOS/cm				
Zinc					Cyanide: Total				
Aluminum					Amenable				
Antimony					Free				
Arsenic					Reactive				
Barium					Flash Point °F				
Beryllium					Fluoride				
Calcium					Hardness (CaCO ₃)				
Iron					MBAS				
Magnesium					Nitrogen: Nitrate (N)				
Manganese					Nitrite (N)				
Mercury					Ammonia (N)				
Potassium					T Kjeldahl (N)				
Selenium					Oil & Grease				
Silicon					Oxygen Demand: BOD ₅				
Sodium					COD				
Thallium					pH				
Tin					Phenols				
					Phosphorus: Total (P)				
					Residue: Total				
					Total Volatile				
					Suspended				
					Volatile Suspended				
					Sulfate				
					Sulfide: Total				
					Reactive				

ug/L = micrograms/Liter(ppb) • mg/L = milligrams/Liter(ppm) • mg/kg = milligrams/kilogram • > = greater than • < = less than (below detection)

Sampled By: Client xx KWLI Auto Sampler Other

KUDUKIS WASTEWATER LABS, INC.

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ANALYSIS CERTIFIED BY: John Ondo

JO Laboratory Manager

Client ENVISAGE ENVIRONMENTAL
 P.O. BOX 152
 RICHFIELD, OHIO 44286
 ATTN: AUDREY

Report Date: March 20, 1991
 P.O. #: _____
 Client: _____
 Sample Received: 3/14/91
 Lab Sample #: 1-0756-0758

Sample I.D.: #1 - EEI - OR3-1, 3/12/91
 #2 - EEI - OR3-2, 3/12/91
 #3 - EEI - OR3-3, 3/12/91

ANALYSIS	#1	#2	#3	#4	ANALYSIS	#1	#2	#3	#4
Metals:	Total mg/L	Total mg/L	Total mg/L	—	Acidity (CaCO ₃)				
Cadmium					Alkalinity (CaCO ₃)				
Chromium: Total					Bacteria:				
Hexavalent					F. Coliform/100ml				
Trivalent					T. Coliform/100ml				
Copper					Total Plate Count/1ml				
Lead	0.58	0.05	<0.05		Chloride				
Nickel					Chlorine: T. Res.				
Silver					Conductivity - uMHOS/cm				
Zinc					Cyanide: Total				
Aluminum					Amenable				
Antimony					Free				
Arsenic					Reactive				
Barium					Flash Point °F				
Beryllium					Fluoride				
Calcium					Hardness (CaCO ₃)				
Iron					MBAS				
Magnesium					Nitrogen: Nitrate (N)				
Manganese					Nitrite (N)				
Mercury					Ammonia (N)				
Potassium					T Kjeldahl (N)				
Selenium					Oil & Grease				
Silicon					Oxygen Demand: BOD ₅				
Sodium					COD				
Thallium					pH				
Tin					Phenols				
					Phosphorus: Total (P)				
					Residue: Total				
					Total Volatile				
					Suspended				
					Volatile Suspended				
					Sulfate				
					Sulfide: Total				
					Reactive				

ug/L = micrograms/Liter(ppb) • mg/L = milligrams/Liter(ppm) • mg/kg = milligrams/kilogram • > = greater than • < = less than (below detection)

Sampled By: Client XX KWLI _____ Auto Sampler _____ Other _____

KUDUKIS WASTEWATER LABS, INC.

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ANALYSIS CERTIFIED BY: _____

John Ondo

JO Laboratory Manager

Client ENVISAGE ENVIRONMENTAL
P.O. BOX 152
RICHFIELD, OHIO 44286

ATTN: AUDREY

Report Date: March 18, 1991
P.O. #: _____
Client: _____
Sample Received: 3/14/91
Lab Sample #: 1-0765-0767

Sample I.D.: #1 - EEI - Filter 245
#2 - EEI - Filter 247
#3 - EEI - Filter 252

ANALYSIS	#1	#2	#3	#4	ANALYSIS	#1	#2	#3	#4
Metals:	Total	Total	Total	—	Acidity (CaCO ₃)				
Cadmium					Alkalinity (CaCO ₃)				
Chromium: Total					Bacteria:				
Hexavalent					E. Coliform/100ml				
Trivalent					T. Coliform/100ml				
Copper					Total Plate Count/1ml				
Lead mg/Filter	0.11	0.12	0.19		Chloride				
Nickel					Chlorine: T. Res.				
Silver					Conductivity - uMHOS/cm				
Zinc					Cyanide: Total				
Aluminum					Amenable				
Antimony					Free				
Arsenic					Reactive				
Barium					Flash Point °F				
Beryllium					Fluoride				
Calcium					Hardness (CaCO ₃)				
Iron					MBAS				
Magnesium					Nitrogen: Nitrate (N)				
Manganese					Nitrite (N)				
Mercury					Ammonia (N)				
Potassium					T Kjeldahl (N)				
Selenium					Oil & Grease				
Silicon					Oxygen Demand: BOD ₅				
Sodium					COD				
Thallium					pH				
Tin					Phenols				
					Phosphorus: Total (P)				
					Residue: Total				
					Total Volatile				
					Suspended				
					Volatile Suspended				
					Sulfate				
					Sulfide: Total				
					Reactive				

ug/L = micrograms/Liter(ppb) • mg/L = milligrams/Liter(ppm) • mg/kg = milligrams/kilogram • > = greater than • < = less than (below detection)

Sampled By: Client xx KWLI _____ Auto Sampler _____ Other _____

KUDUKIS WASTEWATER LABS, INC.

2779 BROADWAY AVENUE, CLEVELAND, OHIO 44115

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ANALYSIS CERTIFIED BY: John Ondo

JO Laboratory Manager

Client ENVISAGE ENVIRONMENTAL
P.O. BOX 152
RICHFIELD, OHIO 44286
ATTN: AUDREY

Report Date: March 20, 1991
P.O. #: _____
Client: _____
Sample Received: 3/14/91
Lab Sample #: 1-0747-0749

Sample I.D.: #1 - EEI - NR3-1, 3/12/91
#2 - EEI - NR3-2, 3/12/91
#3 - EEI - NR3-3, 3/12/91

ANALYSIS	#1	#2	#3	#4	ANALYSIS	#1	#2	#3	#4
Metals:	Total mg/L	Total mg/L	Total mg/L	—	Acidity (CaCO ₃)				
Cadmium					Alkalinity (CaCO ₃)				
Chromium: Total					Bacteria:				
Hexavalent					F. Coliform/100ml				
Trivalent					T. Coliform/100ml				
Copper					Total Plate Count/1ml				
Lead	0.24	<0.05	0.09		Chloride				
Nickel					Chlorine: T. Res.				
Silver					Conductivity - uMHOS/cm				
Zinc					Cyanide: Total				
Aluminum					Amenable				
Antimony					Free				
Arsenic					Reactive				
Barium					Flash Point °F				
Beryllium					Fluoride				
Calcium					Hardness (CaCO ₃)				
Iron					MBAS				
Magnesium					Nitrogen: Nitrate (N)				
Manganese					Nitrite (N)				
Mercury					Ammonia (N)				
Potassium					T Kjeldahl (N)				
Selenium					Oil & Grease				
Silicon					Oxygen Demand: BOD ₅				
Sodium					COD				
Thallium					pH				
Tin					Phenols				
					Phosphorus: Total (P)				
					Residue: Total				
					Total Volatile				
					Suspended				
					Volatile Suspended				
					Sulfate				
					Sulfide: Total				
					Reactive				

ug/L = micrograms/Liter(ppb) • mg/L = milligrams/Liter(ppm) • mg/kg = milligrams/kilogram • > = greater than • < = less than (below detection)

Sampled By: Client xx KWLI _____ Auto Sampler _____ Other _____

KUDUKIS WASTEWATER LABS, INC.

2779 BROADWAY AVENUE, CLEVELAND, OHIO 44115

(216) 696 - 0280 FAX: (216) 696 - 6831

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ANALYSIS CERTIFIED BY: John Ondo

JO Laboratory Manager

Client ENVISAGE ENVIRONMENTAL
P.O. BOX 152
RICHFIELD, OHIO 44286
ATTN: AUDREY

Report Date: March 20, 1991

P.O. #:

Client:

Sample Received: 3/14/91

Lab Sample #: 1-0741-0743

Sample I.D.: #1 - E.E.I - NR1-1 3-12-91
#2 - E.E.I - NR1-2 3-12-91
#3 - E.E.I - NR1-3 3/12/91

MAR 25 1991

ANALYSIS	#1	#2	#3	#4	ANALYSIS	#1	#2	#3	#4
Metals:	Total mg/L	Total mg/L	Total mg/L	—	Acidity (CaCO ₃)				
Cadmium					Alkalinity (CaCO ₃)				
Chromium: Total					Bacteria:				
Hexavalent					F. Coliform/100ml				
Trivalent					T. Coliform/100ml				
Copper					Total Plate Count/1ml				
Lead	0.13	0.06	0.15		Chloride				
Nickel					Chlorine: T. Res.				
Silver					Conductivity - uMHOS/cm				
Zinc					Cyanide: Total				
Aluminum					Amenable				
Antimony					Free				
Arsenic					Reactive				
Barium					Flash Point °F				
Beryllium					Fluoride				
Calcium					Hardness (CaCO ₃)				
Iron					MBAS				
Magnesium					Nitrogen: Nitrate (N)				
Manganese					Nitrite (N)				
Mercury					Ammonia (N)				
Potassium					T Kjeldahl (N)				
Selenium					Oil & Grease				
Silicon					Oxygen Demand: BOD ₅				
Sodium					COD				
Thallium					pH				
Tin					Phenols				
					Phosphorus: Total (P)				
					Residue: Total				
					Total Volatile				
					Suspended				
					Volatile Suspended				
					Sulfate				
					Sulfide: Total				
					Reactive				

ug/L = micrograms/Liter(ppb) • mg/L = milligrams/Liter(ppm) • mg/kg = milligrams/kilogram • > = greater than • < = less than (below detection)

Sampled By: Client xx KWLI Auto Sampler Other

KUDUKIS WASTEWATER LABS, INC.

2779 BROADWAY AVENUE, CLEVELAND, OHIO 44115

(216) 696 - 0280 FAX: (216) 696 - 6831

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ANALYSIS CERTIFIED BY:

John Ondo

JO Laboratory Manager

Client

ENVISAGE ENVIRONMENTAL
P.O. BOX 152
RICHFIELD, OHIO 44286
ATTN: AUDREY

Report Date: March 20, 1991

P.O. #: _____

Client: _____

Sample Received: 3/14/91

Lab Sample #: 1-0744-0746

Sample I.D.: #1 - EEI - NR2-1, 3/12/91
#2 - EEI - NR2-2, 3/12/91
#3 - EEI - NR2-3, 3/12/91

ANALYSIS	#1	#2	#3	#4	ANALYSIS	#1	#2	#3	#4
Metals:	Total mg/L	Total mg/L	Total mg/L	—	Acidity (CaCO ₃)				
Cadmium					Alkalinity (CaCO ₃)				
Chromium: Total					Bacteria:				
Hexavalent					F. Coliform/100ml				
Trivalent					T. Coliform/100ml				
Copper					Total Plate Count/1ml				
Lead	0.13	0.10	0.15		Chloride				
Nickel					Chlorine: T. Res.				
Silver					Conductivity - uMHOS/cm				
Zinc					Cyanide: Total				
Aluminum					Amenable				
Antimony					Free				
Arsenic					Reactive				
Barium					Flash Point °F				
Beryllium					Fluoride				
Calcium					Hardness (CaCO ₃)				
Iron					MBAS				
Magnesium					Nitrogen: Nitrate (N)				
Manganese					Nitrite (N)				
Mercury					Ammonia (N)				
Potassium					T Kjeldahl (N)				
Selenium					Oil & Grease				
Silicon					Oxygen Demand: BOD ₅				
Sodium					COD				
Thallium					pH				
Tin					Phenols				
					Phosphorus: Total (P)				
					Residue: Total				
					Total Volatile				
					Suspended				
					Volatile Suspended				
					Sulfate				
					Sulfide: Total				
					Reactive				

ug/L = micrograms/Liter(ppb) • mg/L = milligrams/Liter(ppm) • mg/kg = milligrams/kilogram • > = greater than • < = less than (below detection)

Sampled By: Client xx KWLI _____ Auto Sampler _____ Other _____

KUDUKIS WASTEWATER LABS, INC.

2779 BROADWAY AVENUE, CLEVELAND, OHIO 44115

(216) 696 - 0280 FAX: (216) 696 - 6831

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ANALYSIS CERTIFIED BY: John Ondo

JO Laboratory Manager

Client ENVISAGE ENVIRONMENTAL
P.O. BOX 152
RICHFIELD, OHIO 44286

ATTN: AYDREY

Report Date: March 20, 1991
P.O. #: _____
Client: _____
Sample Received: 3/14/91
Lab Sample #: 1-0759-0761

Sample I.D.: #1 - EEI - 80% IPA Blank, 3/12/91
#2 - EEI - 0.1 NHNO₃ Blank, 3/12/91
#3 - EEI - 3% H₂O₂ Blank, 3/12/91

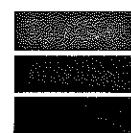
ANALYSIS	#1	#2	#3	#4	ANALYSIS	#1	#2	#3	#4
Metals:	Total mg/L	Total mg/L	Total mg/L	—	Acidity (CaCO ₃)				
Cadmium					Alkalinity (CaCO ₃)				
Chromium: Total					Bacteria:				
Hexavalent					F. Coliform/100ml				
Trivalent					T. Coliform/100ml				
Copper					Total Plate Count/1ml				
Lead	<0.05	<0.05	<0.05		Chloride				
Nickel					Chlorine: T. Res.				
Silver					Conductivity - uMHOS/cm				
Zinc					Cyanide: Total				
Aluminum					Amenable				
Antimony					Free				
Arsenic					Reactive				
Barium					Flash Point °F				
Beryllium					Fluoride				
Calcium					Hardness (CaCO ₃)				
Iron					MBAS				
Magnesium					Nitrogen: Nitrate (N)				
Manganese					Nitrite (N)				
Mercury					Ammonia (N)				
Potassium					T Kjeldahl (N)				
Selenium					Oil & Grease				
Silicon					Oxygen Demand: BOD ₅				
Sodium					COD				
Thallium					pH				
Tin					Phenols				
					Phosphorus: Total (P)				
					Residue: Total				
					Total Volatile				
					Suspended				
					Volatile Suspended				
					Sulfate				
					Sulfide: Total				
					Reactive				

ug/L = micrograms/Liter(ppb) • mg/L = milligrams/Liter(ppm) • mg/kg = milligrams/kilogram • > = greater than • < = less than (below detection)

Sampled By: Client xx KWLI _____ Auto Sampler _____ Other _____

EMISSION SAMPLING

EQUIPMENT SPECIFICATIONS



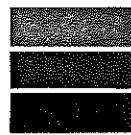
**Envisage
Environmental
Incorporated**

P.O. Box 152 Richfield, Ohio 44286
Phone (216) 526-0990

Equipment and Specifications
U.S.E.P.A. Reference methods 1-5

Control Unit (Meter Box)	Equipment Designation	
<input type="checkbox"/> Envisage Environmental Inc.	Control Unit #'s MB- 08 & 09	
<input type="checkbox"/> Anderson Samplers	Control Unit #'s MB- 01 - 02	
<input checked="" type="checkbox"/> Remanufactured R.A.C.	Control Unit #'s MB- 03 - 07	
Sample Box		
<input checked="" type="checkbox"/> E.E.I.	SB- 01, 02 & 05 - 07	
<input type="checkbox"/> Remanufactured R.A.C.	SB- 03 & 04	
<input type="checkbox"/> E.E.I. Special Design	SB- 08 - 11	
Impingers - per sample train (each set changed for each test run)		
<input checked="" type="checkbox"/> E.E.I.	3 Modified Smith Greenburg type	
<input checked="" type="checkbox"/> E.E.I.	1 Smith Greenburg type	
Probes	length	Lining types
<input checked="" type="checkbox"/> E.E.I.	2 foot	SS, PYREX, QUARTZ
<input checked="" type="checkbox"/> E.E.I.	3 foot	SS, PYREX, QUARTZ, TEFILON
<input type="checkbox"/> E.E.I.	5 foot	SS, PYREX, QUARTZ, TEFILON
<input type="checkbox"/> E.E.I.	3 foot	SS, PYREX, TEFILON
<input type="checkbox"/> E.E.I.	7 foot	SS, PYREX, TEFILON
<input type="checkbox"/> E.E.I.	10 foot	SS, PYREX, TEFILON
<input type="checkbox"/> E.E.I.	12 foot	SS, PYREX, TEFILON
<input type="checkbox"/> E.E.I.	15 foot	SS, PYREX, TEFILON
<input type="checkbox"/> E.E.I.	24 foot	SS, TEFILON
Temperature Sensors	Equipment Designation	
<input type="checkbox"/> Omega Engineering	PY- 01 & 02	
<input type="checkbox"/> Thermo Electric	PY- 03 - 08	
<input checked="" type="checkbox"/> Fluke 51	PY- 01 - 02 - 03 - 04 - 05	
<input type="checkbox"/> Fisher Scientific	Mercury Thermometer	
<input type="checkbox"/> Fisher Scientific	Bimetallic Thermometer	
Pressure Gauges	Type	
<input checked="" type="checkbox"/> Dwyer Incline Manometer	Oil, 0 - 10 inch water	
<input type="checkbox"/> Dwyer Magnehelic	Magnetic/Mechanical 0 - 1 inch water	
<input type="checkbox"/> Dwyer Magnehelic	Magnetic/Mechanical 0 - 10 inch water	
<input type="checkbox"/> Dwyer "U" Tube Manometer	Mercury, 36 inches	
<input type="checkbox"/> Dwyer "U" Tube Manometer	Water, 72 inches	
<input type="checkbox"/> Dwyer Microtector (Micro - Manometer)	Water, 0 - 1 inches of water	
Chemicals and Reagents		
<input checked="" type="checkbox"/> Water	Deionized/ Distilled	
<input checked="" type="checkbox"/> Acetone	Reagent Grade (<0.001% residual)	
<input checked="" type="checkbox"/> Silica Gel	6 - 16 Mesh	
<input checked="" type="checkbox"/> Stopcock Grease	Acetone- Insoluble & Heat Resistant	

CALIBRATION SECTION



**Envisage
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P.O. Box 152 Richfield, Ohio 44286
Phone (216) 526-0990

"S" TYPE PITOT TUBE CALIBRATION

"S" Type Pitot Tube (Probe) # # 303 - 3 ft. Probe

Calibration Date: March 4, 1991

$$\frac{C_p}{C_{std}} = \frac{\frac{\frac{\Delta P}{\Delta P_{std}}}{\sqrt{\frac{\Delta P}{P}}}}{\sqrt{\frac{\Delta P}{P}}} \quad (\text{EPA Equation 2-2})$$

where:

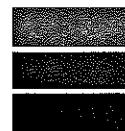
C_p = Coefficient of Type S pitot tube, dimensionless

C_{std} = Coefficient of Standard Pitot Tube (0.99), dimensionless

ΔP_{std} = Velocity head measured by standard pitot tube, inches H_2O

ΔP_p = Velocity head measured by Type S pitot tube, inches H_2O

	ΔP_{std}	ΔP_p	C_p
Side A	0.16	0.22	0.844
Side B	0.16	0.22	0.844
Side A	0.30	0.42	0.837
Side B	0.30	0.42	0.837
Side A	0.52	0.72	0.841
Side B	0.52	0.72	0.841
Average -			0.84



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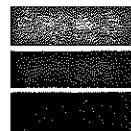
METER BOX CALIBRATION

Meter Box Number: RAC # 4Calibration Date: March 5, 1991

$$Y = \frac{V_t P_b (T_m + 460)}{V_m \left[P_b + \frac{\Delta H}{13.6} \right] (T + 460)}$$

$$\Delta H_0 = \frac{0.0317 \Delta H}{P_b (T_m + 460)} \left[\frac{(T_t + 460) t}{V_t} \right]$$

Delta H (ΔH)	in. H2O	0.5	1.0	3.0	5.0	7.0
Pres. Barometer (P_b)	in. Hg	29.62	29.62	29.62	29.62	29.62
Vol. Meter Box (V_m)	cu. ft.	4.219	6.010	10.400	13.409	15.826
Vol. Test Meter (V_t)	cu. ft.	4.152	5.825	10.045	12.900	15.209
Temp. Meter Box (T_m)	$^{\circ}$ F	88.0	95.7	100.7	106.9	110.2
	$^{\circ}$ R	548.0	555.7	560.7	566.9	570.2
Temp. Test Meter (T_t)	$^{\circ}$ F	78.0	78.0	78.0	78.0	78.0
	$^{\circ}$ R	538.0	538.0	538.0	538.0	538.0
Time (t)	minutes	10.0	10.0	10.0	10.0	10.0
METER FACTOR (Y)		1.001	0.999	0.999	1.001	1.001
- Average				1.00		
METER COEFFICIENT (ΔH_0)		1.640	1.643	1.643	1.642	1.644
- Average				1.64		



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METER BOX CALIBRATION

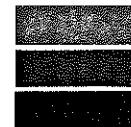
Meter Box Number: RAC # 6

Calibration Date: January 29, 1991

$$Y = \frac{V_t P_b (T_m + 460)}{V_m \left[\frac{P_b + \frac{\Delta H}{13.6}}{(T_m + 460)} \right]}$$

$$\Delta H_e = \frac{0.0317 \Delta H}{P_b (T_m + 460)} \left[\frac{(T_t + 460) t}{V_t} \right]$$

Delta H (ΔH)	in. H2O	0.5	1.0	3.0	5.0	7.0
Pres. Barometer (P_b)	in. Hg	29.45	29.45	29.45	29.45	29.45
Vol. Meter Box (V_m)	cu. ft.	4.018	5.689	9.878	12.697	14.969
Vol. Test Meter (V_t)	cu. ft.	3.963	5.589	9.613	12.391	14.646
Temp. Meter Box (T_m)	°F	87.0	90.0	97.8	99.4	100.6
	°R	547.0	550.0	557.8	559.4	560.6
Temp. Test Meter (T_t)	°F	79.0	79.0	79.0	79.0	79.0
	°R	539.0	539.0	539.0	539.0	539.0
Time (t)	minutes	10.0	10.0	10.0	10.0	10.0
METER FACTOR (Y)		1.000	1.000	1.000	1.000	1.000
- Average				1.00		
METER COEFFICIENT (ΔH_e)		1.820	1.820	1.820	1.820	1.820
- Average				1.82		



**Envisage
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"S" TYPE PITOT TUBE CALIBRATION

"S" Type Pitot Tube (Probe) # 301 - 3 ft Probe

Calibration Date: January 14, 1991

$$\frac{C}{P} = \frac{C_{std}}{\sqrt{\frac{^P_{std}}{^P_p}}} \quad (\text{EPA Equation 2-2})$$

where:

C = Coefficient of Type S pitot tube, dimensionless

p

C_{std} = Coefficient of Standard Pitot Tube (0.99), dimensionless

std

$^P_{std}$ = Velocity head measured by standard pitot tube, inches H O₂

P_p = Velocity head measured by Type S pitot tube, inches H O₂

p

	$^P_{std}$	P_p	$\frac{C}{P}$
Side A	0.15	0.21	0.837
Side B	0.15	0.21	0.837
Side A	0.25	0.35	0.837
Side B	0.25	0.35	0.837
Side A	0.40	0.55	0.844
Side B	0.40	0.55	0.844
Average -			0.84

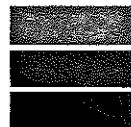
NOZZLE DIAMETER CALIBRATION

I.D. of nozzles are checked periodically by inside micrometer on at least 12 different diameters. If deviation exceeds +0.001" on an average or 0.002" maximum, nozzle is reworked. Sharpening occurs after each test.

CALIBRATION FREQUENCY

The frequency of calibration is dictated by the Federal Register, Volume 42, Number 160, August 18, 1977. The regulations state that you must "use methods and equipment which have been approved by the Administrator to calibrate the orifice meter, pitot tube, dry gas meter, and probe heater. Recalibrate after each test".

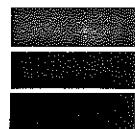
The methods of calibration are determined from "Maintenance, Calibration, and Operation of Isokinetic Source Sampling Equipment," published by the U.S. EPA Office of Air Program Publications APTD-0576. Per the above listed regulations, the equipment was checked after the stack test and the values of Y, Cp (Test) and nozzle diameter had not appreciably changed from the acceptable tolerances.



**Envisage
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P.O. Box 152 Richfield, Ohio 44286
Phone (216) 526-0990

FIELD DATA SHEETS



**Envisage
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Incorporated**

P.O. Box 152 Richfield, Ohio 44286
Phone (216) 526-0990

PLANT Master Metals
DATE 3/12/91
SAMPLING LOCATION New Outlet Furnace
SAMPLE TYPE M. -6
OPERATOR JD NG
AMBIENT TEMPERATURE 40's
BAROMETRIC PRESSURE
STATIC PRESSURE - 7.2
HEATER BOX SETTING 24°F ± 25°F

FIELD DATA



PROBE LENGTH & TYPE 3Drex
NOZZLE I.D. 3/16"
ASSUMED MOISTURE % 8
METER BOX NUMBER 4
METER AH 1.04
C FACTOR .90
PITOT CORRECTION FACTOR .84
PRE-TEST LEAK CHECK 0 CFM @ 17 °F
POST-TEST LEAK CHECK 0 CFM @ 9 °F

SCHEMATIC OF TRAVERSE POINT LAYOUT
READ AND RECORD ALL DATA EVERY 5 MINUTES

#2 0 @ 41"
#4 0 @ 4.8"

PAGE 1 OF _____

Run #1 Case #3

TRAVERSE POINT NUMBER	ELAPSED SAMPLING TIME min.	GAS METER READING	VELOCITY HEAD	ORIFICE PRESSURE DIFFERENTIAL	STACK TEMPERATURE	GAS METER TEMPERATURE	PUMP VACUUM	FILTER HOLDER TEMP.	IMPINGER TEMP.
						INLET	OUTLET		
1	0/0934	314.000	1.80	1.82	115	94	74	5.0	260
5		317.72	1.80	1.82	116	106	76	6.0	265
10		321.84	1.90	1.371	115	118	78	6.0	259
14	15	320.10	1.85	1.80	114	125	80	6.0	261
5	20	330.20	1.80	1.342	112	128	83	6.0	247
10	25	334.36	1.85	1.360	110	127	85	6.0	245
1	30	338.67	2.30	1.517	110	111	90	7.0	261
2	35	343.27	2.10	1.449	111	130	91	7.0	270
3	40	347.80	1.90	1.379	112	135	94	6.0	268
4	45	352.09	1.80	1.342	112	136	94	6.0	258
5	50	356.42	1.70	1.304	112	136	96	6.0	266
6	55	360.35	1.60	1.205	109	134	97	5.5	252
100/1040		364.295							38
		30.295	1.365	2.11	112.33	105.04		N. Norman DAPK	
								31/12/91	
Length check OK									

#2 0 @ 37"

KUN #3 CASE #77

pre 0 @ 15"

F2 08 5.0"
F4 08 5.0"

PLANT MASTER METALS - France DATE 3/12/91 LEAK CHECK ^{post} 0 CFM@ 7 "Hg PAGE 3 OF 1

post pilots 02 0 @ 45"
04 12 @ 52"

८७

KUN #L CASE # 7

pre O @ 18"

"2 O @ 18"
"4 O @ 5.0"PLANT MASTER Meters FURNACE
DATE 3/12/91 LEAK CHECK post 0 CFM@ 11 "Hg PAGE 2 OF _____

TRaverse Point Number	Elapsed Sampling Time min.	GAS METER READING	VELOCITY HEAD	ORIFICE PRESSURE DIFFERENTIAL	STACK TEMPERATURE	GAS METER TEMPERATURE INLET	GAS METER TEMPERATURE OUTLET	PUMP VACUUM	FILTER HOLDER TEMP.	IMPINGER TEMP.	
1	01 1121	364.600	1.60	1.265	108	108	103	5.0	264	35	
2	5	368.62	2.10	1.409	109	132	102	4.0	271	35	
3	10	373.22	1.90	1.378	108	142	105	5.0	268	36	
4	15	377.50	1.75	1.323	111	146	106	5.5	266	36	
5	20	381.70	1.75	1.323	104	148	108	5.5	235	36	
10	25	385.92	1.70	1.304	102	147	107	5.0	237	38	
1	30	390.04	1.70	1.304	101	136	109	5.0	258	37	
2	35	394.20	2.10	1.409	110	142	109	4.0	263	37	
3	40	398.74	2.05	1.432	100	147	108	5.5	259	37	
4	45	403.20	2.00	1.414	2.20	99	147	109	5.5	249	37
5	50	407.77	1.95	1.410	2.15	96	150	111	5.5	250	38
10	55	412.25	2.30	1.517	2.50	95	151	112	6.0	252	38
<u>60 / 12240</u>											
<u>47113</u>											
<u>52.53</u>											
<u>1.38</u>											
<u>2.075</u>											
<u>103.58</u>											
<u>124.46</u>											
<u>Leak check OK</u>											
<u>i. Sciaman PARC</u>											
<u>3/12/91</u>											

post 12 O @ 4.8"
pbits ft O @ 4.0"

TC

FIELD DATA

PLANT Master Metals

DATE 3-12-91

SAMPLING LOCATION Fugitive Baghouse

SAMPLE TYPE 1-6

OPERATOR RH SN

AMBIENT TEMPERATURE 50°

BAROMETRIC PRESSURE

STATIC PRESSURE ~~1.09~~ + 1.9

HEATER BOX SETTING 245 ± 25°

36"

diameter

PROBE LENGTH & TYPE 3 ft pyrex

NOZZLE I.D. ~~1.5"~~ 0.187"

ASSUMED MOISTURE % 5

METER BOX NUMBER 6

METER AH @ 1.02

C FACTOR 1.0

PITOT CORRECTION FACTOR .84

PRE-TEST LEAK CHECK 0 CPM@ 1 "Hg

POST-TEST LEAK CHECK 0 CPM@ 9 "Hg

Pitot Tubes ± 2 1/2 4.3

84 ± 3

SCHEMATIC OF TRAVERSE POINT LAYOUT

READ AND RECORD ALL DATA EVERY 5 MINUTESPAGE OF

Run 1 CASE #8

TRAVERSE POINT NUMBER	ELAPSED SAMPLING TIME min.	GAS METER READING	VELOCITY HEAD	ORIFICE PRESSURE DIFFERENTIAL	STACK TEMPERATURE	GAS METER TEMPERATURE	PUMP VACUUM	FILTER HOLDER TEMP.	IMPINGER TEMP.
					INLET	OUTLET			
1	011 940	772.40	2.9	1.70	54	90	5.0	244	36
2	5	771.99	2.8	1.67	54	80	5.0	266	36
3	10	781.76	2.9	1.70	54	88	5.5	270	36
4	15	786.87	2.7	1.64	54	90	5.0	244	36
5	20	791.48	2.5	1.58	54	72	5.0	272	36
6	25	794.16	2.7	1.64	54	72	6.0	264	36
7	30	801.34	2.6	1.61	54	72	5.0	273	38
8	35	805.82	2.7	1.64	54	73	6.0	262	38
9	40	810.60	2.7	1.64	54	74	6.0	238	38
10	45	815.45	2.9	1.70	54	76	7.0	273	38
11	50	820.44	2.9	1.70	54	76	7.0	265	38
12	55	825.47	2.8	1.67	55	76	6.0	270	38
	60/11045	832.345							
		59.945	1.663	3.10	54.17	here checked OK			
						N. Kanan	DAPC		
							511.191		
							78.79		

Post P. ht ± 0 @ 4.0

84.0 @ 3.4

(C)
(D)

TRaverse Point Number	Elapsed Sampling Time min.	Manometer Readings				Velocity Head				Orifice Pressure Differential				stack Temperature	gas meter inlet	outlet	Pump vacuum	filter holder	impinger temp.
		CH 2	CH 4	CH 8	CH 14	P1	P2	P3	P4	diff.	P1	P2	P3	P4					
1	0/11/30 0	830.70	2.9	1.76	3.3	5.6	7.2	6.0	2.35	3.3									
2	10	635.75	2.9	1.70	3.3	5.7	6.6	4.0	2.42	3.9									
3	10	840.82	2.9	1.67	3.1	5.7	7.2	5.5	2.47	3.8									
4	15	615.75	3.0	1.73	3.5	5.7	6.4	4.5	2.5D	3.4									
5	20	625.03	2.8	1.67	3.1	5.7	9.8	7.8	5.5	2.64	3.8								
6	25	635.90	2.7	1.64	3.0	5.7	9.1	8.0	5.5	2.60	4.0								
7	30	620.76	2.9	1.67	3.1	5.7	8.0	5.5	2.63	4.0									
8	35	625.42	2.8	1.67	3.1	5.7	9.3	8.0	5.5	2.70	4.2								
9	40	612.55	2.9	1.70	3.3	5.7	9.5	8.0	6.0	2.65	4.2								
10	45	617.5.60	2.7	1.64	3.0	5.7	9.6	8.0	5.0	2.72	4.2								
11	50	620.86	2.5	1.58	2.9	5.7	9.6	7.1	5.0	2.72	4.2								
12	55	625.42	2.5	1.58	2.9	5.7	9.5	8.0	5.0	2.59	4.2								
13	60/12/24	625.75	2.5	1.58	2.9	5.7	9.5	8.0	5.0	2.59	4.2								
14		620.15																	

Depth 1 foot
 Room 74°C
 3/12/91

11.7 ft 4.2 0@ 3.4 Post 22.0 3.1
 11.5 4.2 0@ 3.4 Post 22.0 3.1
 11.5 4.2 0@ 3.4 Post 22.0 3.1

PLANT Master Monitor Tramatic

DATE 3/2/61

Run 3 Case 20

PAGE 3 OF

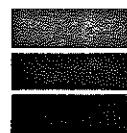
LEAK CHECK @ CFM 17 "Hg

TRAVERSE POINT NUMBER	ELAPSED SAMPLING TIME min.	GAS METER READING	VELOCITY HEAD	ORIFICE PRESSURE DIFFERENTIAL	STACK TEMPERATURE	GAS METER INLET	GAS METER OUTLET	PUMP VACUUM	FILTER HOLDER TEMP.	IMPINGER TEMP.
1	01 1652	890.60	2.9	1.70	5.3	50	70	5.0	291	44
2	045.53	2.6	1.67	3.1	57	60	72	5.0	245	44
3	500.43	3.0	1.73	3.5	56	66	72	6.0	250	44
4	501.72	2.8	1.67	5.1	57	60	73	5.5	248	44
5	510.68	2.6	1.61	2.65	57	62	74	4.5	253	44
6	515.56	2.5	1.58	2.9	57	64	74	4.0	257	44
7	520.41	2.7	1.64	3.0	56	64	73	5.0	246	44
8	525.44	3.07	1.64	3.0	57	60	71	5.0	236	44
9	530.21	2.8	1.67	3.1	56	62	75	5.0	265	46
10	535.20	2.9	1.70	3.3	57	72	74	6.5	36.5	46
11	540.36	2.9	1.70	3.3	57	62	71	6.5	262	45
12	545.63	2.8	1.67	3.1	56	63	75	6.0	259	46
13	1356 61-0.55									
	59.95	1.665	3.14	56	80.67	Leach check	cK			
						N. Farnan	DMSL			
						3/12/61				

R.644 #2 04 3.8 Not #2 out 3.1
2.0 2.1 4.2 4.4 0 of 3.2

EMISSIONS SAMPLING

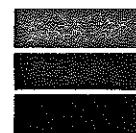
NOMENCLATURE



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PARTICULATE SAMPLING NOMENCLATURE

- A = Cross sectional area of stack or duct, ft².
- A_n = Cross sectional area of nozzle, ft².
- B_{ws} = Water vapor in gas stream, proportion by volume.
- C = Nomograph correction factor, dimensionless.
- C_p = Pitot tube coefficient, dimensionless.
- C_s = Concentration of particulate matter in gas stream, dry basis-corrected to standard conditions, gr/dscf.
- D_n = Nominal diameter of probe nozzle tip, inches.
- E = Particulate Emission Rate, lb/hr.
- ΔH = Average pressure differential across orifice, in. H₂O.
- ΔH₀ = Orifice meter calibration factor, in. H₂O.
- I = Percent of Isokinetic sampling, %.
- K_p = Pitot tube constant, 85.49 $\frac{\text{ft}}{\text{sec}} \left[\frac{(\text{lb/lb-mole})(\text{in.Hg})}{(\text{R})(\text{in.H}_2\text{O})} \right]$
- M_d = Molecular weight of gas, dry basis, lb/lb-mole.
- M_n = Total amount of particulate matter collected, g.
- M_s = Molecular weight of gas, wet basis, lb/lb-mole.

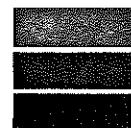


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Particulate Sampling Nomenclature - continued

- M_w = Molecular weight of water, 18 lb/lb-mole.
- P_{bar} = Barometric Pressure, in. Hg.
- P_g = Pressure differential from gas stream to atmosphere,
(static pressure) in. H_2O .
- P_s = Absolute gas stream pressure, ($P_{bar} + P_g/13.6$) in.Hg.
- P_{std} = Absolute pressure at standard conditions, 29.92 in. Hg.
- P_w = Density of water, 0.0022 lb/ml.
- \overline{P}_{avg} = Average of the square roots of the velocity head readings,
($\sqrt{\overline{p}}$) (in. H_2O).
- Q = Volumetric flow rate at gas stream conditions, A.C.F.M.
- Q_{sd} = Dry volumetric gas flow rate corrected to standard
conditions, S.C.F.M.
- R = Ideal gas constant, 21.85 in. Hg-ft³/°R-lb-mole.
- t = Total sampling time, minutes.
- T_m = Average dry gas meter temperature, °R.
- T_s = Average absolute gas stream temperature, °R.
- T_{std} = Standard absolute temperature, 528° Rankine.
- V_{lc} = Volume of water collected in impingers & silica gel, ml.



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Particulate Sampling Nomenclature - continued

V_m = Volume of gas sample measured at meter box (meter conditions), ft³

$V_{m(std)}$ = Volume of gas sample measured at meter box (corrected to standard conditions), ft³.

V_s = Average gas stream velocity, ft/sec.

$V_{w(std)}$ = Volume of water vapor in gas sample (standard conditions) ft³.

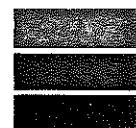
13.6 = Specific gravity of mercury (Hg).

% CO₂ = Percent by volume of CO₂ in gas stream (dry basis).

% O₂ = Percent by volume of O₂ in gas stream (dry basis).

% CO = Percent by volume of CO in gas stream (dry basis).

% N₂ = Percent by volume of N₂ in gas stream (dry basis).

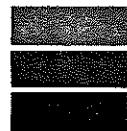


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EMISSION SAMPLING

CALCULATIONS



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- 1) Volume of dry gas sampled through meter box at standard conditions,

$$V_{m(\text{std})} = V_m \left[\frac{T_{\text{std}}}{T_m} \right] \left[\frac{P_b + \frac{\Delta H}{13.6}}{P_{\text{std}}} \right]$$

(EPA Equation 5-1)

Where:

$V_{m(\text{std})}$ = Volume of gas sample measured at meter box (corrected to standard conditions), ft³.

V_m = Volume of gas sample measured at meter box (meter conditions), ft³.

T_{std} = Standard absolute temperature, 528° Rankine.

T_m = Average dry gas meter temperature, °R.

P_{bar} = Barometric Pressure, in. Hg.

ΔH = Average pressure differential across orifice, in. H₂O.

13.6 = Specific gravity of mercury (Hg).

P_{std} = Absolute pressure at standard conditions, 29.92 in. Hg.

Example: Run 1

$$V_m = 59.95 \text{ ft}^3$$

$$T_m = 538.8 \text{ } ^\circ\text{R}$$

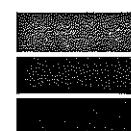
$$\Delta H = 3.10 \text{ in. H}_2\text{O}$$

$$P_{\text{bar}} = 30.18 \text{ in. Hg}$$

$$V_{m(\text{std})} = 59.95 \left[\frac{528.0}{538.8} \right] \left[\frac{30.18 + \frac{3.10}{13.6}}{29.92} \right]$$

$$= 59.95 (0.9799) (1.0163)$$

$$= 59.71 \text{ ft}^3$$



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2) Volume of water vapor collected at standard conditions,

$$V_{w(\text{std})} = V_{1c} \left[\frac{P_w}{M_w} \right] \left[\frac{(R)(T_{\text{std}})}{P_{\text{std}}} \right]$$

(EPA Equation 5-2)

Where:

$V_{w(\text{std})}$ = Volume of water vapor in gas sample (standard conditions) ft^3 .

V_{1c} = Volume of water collected in impingers & silica gel, ml.

P_w = Density of water, 0.0022 lb/ml.

M_w = Molecular weight of water, 18 lb/lb-mole.

R = Ideal gas constant, 21.85 in. Hg-ft 3 /°R-lb-mole.

T_{std} = Standard absolute temperature, 528 ° Rankine.

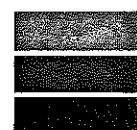
P_{std} = Absolute pressure at standard conditions, 29.92 in. Hg.

Example: Run 1

$$V_{1c} = 16.8 \text{ ml}$$

$$V_{w(\text{std})} = 16.8 \left[\frac{0.0022}{18.0} \right] \left[\frac{(21.85)}{29.92} \frac{(528.0)}{} \right]$$

$$= 0.79 \text{ ft}^3$$



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5) Molecular Weight of gas in gas stream,

$$M_s = M_d (1 - B_{ws}) + M_w (B_{ws})$$

(EPA Equation 2-5)

Where:

M_s = Molecular weight of gas, wet basis, lb/lb-mole.

M_d = Molecular weight of gas, dry basis, lb/lb-mole.

B_{ws} = Water vapor in gas stream, proportion by volume.

M_w = Molecular weight of water, 18 lb/lb-mole.

Example: Run 1

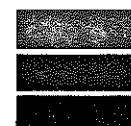
$$M_d = 28.84 \text{ lb/lb-mole}$$

$$B_{ws} = 0.0131$$

$$M_s = 28.84 (1 - 0.0131) + 18 (0.0131)$$

$$= 28.459 + 0.235$$

$$= \underline{\underline{28.69 \text{ lb/lb-mole}}}$$



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6) Average Gas Stream Velocity,

6.)

$$V_s = K_p C_p \sqrt{\frac{T_s}{P_s M_s}}$$

(EPA Equation 2-9)

Where:

V_s = Average gas stream velocity, ft/sec.

K_p = Pitot tube constant, 85.49 $\frac{ft}{sec} \left[\frac{(lb/lb-mole)(in.Hg)}{(R)(in.H_2O)} \right]^{1/2}$

C_p = Pitot tube coefficient, dimensionless.

$\sqrt{P_{avg}}$ = Average of the square roots of the velocity head readings, ($\sqrt{\bar{P}}$) (in. H_2O).

T_s = Average absolute gas stream temperature, oR .

P_s = Absolute gas stream pressure, ($P_{bar} + P_g/13.6$) in.Hg.

P_{bar} = Barometric Pressure, in. Hg.

P_g = Pressure differential from gas stream to atmosphere, (static pressure) in. H_2O .

M_s = Molecular weight of gas, wet basis, lb/lb-mole.

Example: Run 1

C_p = 0.84

$\sqrt{P_{avg}}$ = 1.663 in. $H_2O^{1/2}$

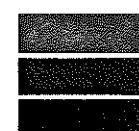
T_s = 514.2 oR

P_s = $P_{bar} + P_g/13.6 = 30.18 + 1.90/13.6 = 30.32$ in.Hg

M_s = 28.69 lb/lb-mole

$$V_s = (85.49)(0.84)(1.663) \sqrt{\frac{514.2}{(30.32)(28.69)}}$$

$= 91.81$ ft/sec



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7) Volumetric Flow Rate at Gas Stream Conditions,

$$Q = A \times V_s \times 60$$

Where:

Q = Volumetric flow rate at gas stream conditions, A.C.F.M.

A = Cross sectional area of stack or duct, ft^2 .

V_s = Average gas stream velocity, ft/sec .

60 = Conversion factor from seconds to minutes.

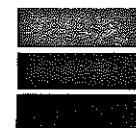
Example: Run 1

$$A = 7.07 \text{ ft}^2$$

$$V_s = 91.81 \text{ ft/sec}$$

$$Q = (7.07) (91.81) 60$$

$$= 38,937 \text{ ACFM}$$



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8) Volumetric Flow Rate at Standard Conditions,

$$Q_{sd} = 60 (1 - B_{ws}) V_s A \left[\frac{T_{std}}{T_s} \right] \left[\frac{P_s}{P_{std}} \right]$$

Where:

(EPA Equation 2-10)

Q_{sd} = Dry volumetric gas flow rate corrected to standard conditions, S.C.F.M.

60 = Conversion factor from seconds to minutes.

B_{ws} = Water vapor in gas stream, proportion by volume.

V_s = Average gas stream velocity, ft/sec.

A = Cross sectional area of stack or duct, ft^2 .

T_{std} = Standard absolute temperature, 528° Rankine.

T_s = Average absolute gas stream temperature, $^{\circ}\text{R}$.

P_s = Absolute gas stream pressure, $(P_{bar} + P_g/13.6)$ in.Hg.

P_{bar} = Barometric Pressure, in. Hg.

P_g = Pressure differential from gas stream to atmosphere, (static pressure) $\text{in.H}_2\text{O}$.

P_{std} = Absolute pressure at standard conditions, 29.92 in. Hg.

Example: Run 1

$$B_{ws} = 0.0131$$

$$V_s = 91.81 \text{ ft/sec}$$

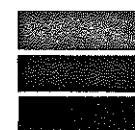
$$A = 7.07 \text{ ft}^2$$

$$T_s = 514.2 \text{ }^{\circ}\text{R}$$

$$P_s = P_{bar} + P_g / 13.6 = 30.18 + 1.90 / 13.6 = 30.32 \text{ in.Hg}$$

$$Q_{sd} = 60 (1 - 0.0131) (91.81) (7.07) \left(\frac{528.0}{514.2} \frac{30.32}{29.92} \right)$$

$$= 39,989 \text{ SCFM}$$



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9) Gas Stream Particulate Concentration,

$$C_s = 15.43 \text{ gr./g} \left[\frac{M_n}{V_{m(\text{std})}} \right]$$

(EPA Equation 5-6)

Where:

C_s = Concentration of particulate matter in gas stream, dry basis-corrected to standard conditions, gr/dscf.

M_n = Total amount of particulate matter collected in probe wash and on filter, g.

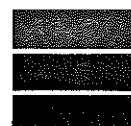
$V_{m(\text{std})}$ = Volume of gas sample measured at meter box (corrected to standard conditions), ft³.

Example: Run 1

$$M_n = (\text{probe}) + (\text{filter}) = 0.0083 + 0.0010 = 0.0093 \text{ g}$$

$$V_{m(\text{std})} = 59.71 \text{ ft}^3$$

$$C_s = 15.43 \left[\frac{0.0093}{59.71} \right] = 0.0024 \text{ gr/dscf}$$



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10) Particulate Emission Rate,

$$E = Q_{sd} C_s \left[\frac{1 \text{ pound}}{7000 \text{ grains}} \right] \left[\frac{60 \text{ minutes}}{1 \text{ hour}} \right]$$

Where:

E = Particulate Emission Rate, lb/hr.

Q_{sd} = Dry volumetric gas flow rate corrected to standard conditions, S.C.F.M.

C_s = Concentration of particulate matter in gas stream, dry basis-corrected to standard conditions, gr/dscf.

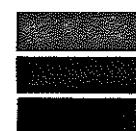
Example: Run 1

$$Q_{sd} = 39,989 \text{ ft}^3$$

$$C_s = 0.0024 \text{ gr/dscf}$$

$$E = (39,989) (0.0024) \left[\frac{60}{7000} \right]$$

$$= 0.82 \text{ lb/hr}$$



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11) Percent of Isokinetic Sampling,

$$I = \frac{100 T_s}{60 A_n} \left[K_3 V_{lc} + \left(\frac{V_m}{T_m} \right) P_{bar} + \frac{\Delta H}{13.6} \right] \quad (\text{EPA Equation 5-7})$$

Where:

I = Percent of Isokinetic sampling, %.

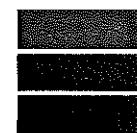
 T_s = Average absolute gas stream temperature, $^{\circ}\text{R}$. K_3 = Constant, 0.002669 in.Hg-ft 3 /ml- $^{\circ}\text{R}$. V_{lc} = Volume of water collected in impingers & silica gel, ml. V_m = Gas sample volume measured at meter box (meter conditions), ft 3 . T_m = Average dry gas meter temperature, $^{\circ}\text{R}$. P_{bar} = Barometric Pressure, in. Hg. ΔH = Average pressure differential across orifice, in. H_2O . t = Total sampling time, minutes. V_s = Average gas stream velocity, ft/sec. P_s = Absolute gas stream pressure, in.Hg. D_n = Nominal diameter of probe nozzle tip, inches. A_n = Cross sectional area of nozzle, ft 2 .

Example: Run 1

 $T_s = 514.2 \text{ } ^{\circ}\text{R}$ $\Delta H = 3.10 \text{ in.} \text{H}_2\text{O}$ $V_{lc} = 16.8 \text{ ml}$ $t = 60.0 \text{ min.}$ $V_m = 59.95 \text{ ft}^3$ $V_s = 91.81 \text{ ft/sec}$ $T_m = 538.8 \text{ } ^{\circ}\text{R}$ $P_s = 30.32 \text{ in.Hg}$ $A_n = 0.0001917 \text{ ft}^2$ $P_{bar} = 30.18 \text{ in.Hg}$

$$I = \frac{514.2 (100)}{60 (0.0001917)} \left[0.002669 (16.8) + \left(\frac{59.95}{538.8} \right) (30.18) + \frac{3.10}{13.6} \right]$$

$$= 91.7 \%$$



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ESTIMATE OF STACK EMISSIONS (FROM NEW BAGHOUSE) WHEN OPERATING THE 3.6-M ROTARY FURNACE,

and Summary of Results of Furnace and Flue Calculations

	Drying		Melting		Reduction	
Operation time during each furnace charging cycle:	1.0	Hr	2.0	Hr	3.0	Hr
Initial temperature for drying, melting, or reduction:	194	°F	590	°F	1445	°F
Temperature at finish of drying, melting, or reduction:	590	°F	1445	°F	1697	°F
Natural gas firing rate:	5.7 MM	Btu/Hr	5.6 MM	Btu/Hr	5.6 MM	Btu/Hr
Average off-gas temperature from the furnace:	1652	°F	1796	°F	2156	°F
Natural gas consumed by the furnace:	5260	Ft ³ /Hr	5245	Ft ³ /Hr	5201	Ft ³ /Hr
Oxygen consumed by the furnace:	8978	Ft ³ /Hr	8953	Ft ³ /Hr	8877	Ft ³ /Hr
Diluted flue gas to the stack						
Flow rate entering stack @ 95 °F :	19,341	ACFM	12,888	ACFM	17,285	ACFM
Flow, DSCFM = 77°F @ 1 atm.:	17,208	DSCFM	11,745	DSCFM	15,822	DSCFM
Moisture content of gas to stack:	4.3%	Vol%	1.9%	Vol%	1.5%	Vol%
Particulates to stack, grains/dscf:	0.0026	g/DSCF	0.0026	g/DSCF	0.0025	g/DSCF
SO ₂ to the stack:	0.51	Lb/Hr	0.12	Lb/Hr	0.029	Lb/Hr
Emissions from Stack (Lb/Hr)						
Tetraethyl lead:	*BDL	Lb/Hr	*BDL	Lb/Hr	*BDL	Lb/Hr
Total chlorine expressed as chloride:	0.023	Lb/Hr	0.016	Lb/Hr	0.021	Lb/Hr
Lead as Pb:	0.13	Lb/Hr	0.085	Lb/Hr	0.11	Lb/Hr
Antimony as Sb:	0.0049	Lb/Hr	0.0032	Lb/Hr	0.0043	Lb/Hr
Arsenic as As:	**ID	Lb/Hr	**ID	Lb/Hr	**ID	Lb/Hr
Barium as Ba:	<0.01	Lb/Hr	<0.01	Lb/Hr	<0.01	Lb/Hr
Silver as Ag:	<0.01	Lb/Hr	<0.01	Lb/Hr	<0.01	Lb/Hr
Cadmium as Cd:	<0.01	Lb/Hr	<0.01	Lb/Hr	<0.01	Lb/Hr
Chromium as Cr:	0.0003	Lb/Hr	0.0002	Lb/Hr	0.0003	Lb/Hr
Mercury, thallium, and beryllium:	***NIF		***NIF		***NIF	

* Below Detectable Limit

** Insufficient data to estimate.

*** Not known to be in feedstock to plant.

Master Metal's 3.6-m SRF Using 100% O₂ to Burner: Simulates 1st Hour After Charge Door is Closed.

Furnace:	Old (3.6-m)	Pb° in Bulln, mt	11.29	CHARGE CYCLE IS FIRST	Bullion	Slag Only	Matte Only
Date:	24-Jul-91	Pb in Feed, mt		PbO			
EndOfPeriod, T	310 degC	Pb,Mat+Slag, mt		Pb°, PbS			
Parameters to Check		Pb in Dust, mt		C°			
Slag FeO =	Target = 10 to 15%	Pb,DrossFloat,mt		CO2			
Basicity =	Target = 1.1 to 1.4	Bullion, mt		FeS			
Viscosity =	Target = 10 to 40	PbRecToBulln		FeO			
(Si+Al)/Na =	1.2	Balanced if small		S			
Slag Ratio of		9.98E-12		Na2S			
(Si+Al)/Na =	1.2	Slag, mt		Na(K)2O			
Slag FeO =	Target = 10 to 15%	Matte, mt		SiO2			
Basicity =	Target = 1.1 to 1.4	Soda/hr to flue	0.045	Ca(Ba)O			
Viscosity =	Target = 10 to 40	Dust, % of feed	2.9%	MgO			
(Si+Al)/Na =	1.2	Molds Output-->		Cr2O3			
Feed(Scoops)	WetFeed,lb	DryFeedMt/hr	Matte + Slag Emulsion, %				
P.CakeTEL(2)	4000.	1.39		Pb			
MM Plate(6)	13600.	6.08		C°			
Dust In (2)	1000.	0.45		FeO			
MM Dross(6)	13000.	5.78		S			
Coke(1/2)	700.	0.31		Na2O			
FeChips(0.35)	500.	0.22		SiO2			
Total Air In	1842.	0.83		Ca(Ba)O			
O2 to Burner	760.	0.34		MgO			
Lenox(0)	0.	0.00		Cr2O3			
NiterDross(0)	0.	0.00		Al2O3			
Glass Sand(2)	2000.	0.89		NaCl			
Anzon(0)	0.	0.0		Sn°, SnO			
Natural Gas	237.4	0.108		Sb°,Sb2S3			
TOTAL	37639.	16.41		Other			
Flux Summary				ZnO			
Soda(2+BH)	3500.	1.586		Other			
Sand (1.5)	2000.	0.907		ZnO			
FeChip(.25)	500.	0.227		Other			
Coke (0.5)	700.	0.317		CuO			
Comments: Moisture is evaporated from charge. The door is closed. There is no fusion or reduction. See Page 2 for gas analyses and data. Oxygen make-up is met with tonnage oxygen (to burner), plus tramp air.				MgCr Brick, kg/hr per mt Bullion	% of S fixed:	99.90%	
				0.0	Coke in matte		
				Al2O3 Brick, Kg/hr per mt Bullion	+ slag, %	"Slag" Basicity	
				0.00			
				Mat.FeS/Na2S	0.50	Distribution Factors	
				Mat.FeO Contr.	10	Sn(Bullion):	0.00
				% Excess Coke	20.0%	Sb(Bullion):	0.00
				PbTo"slag"mt		Cu(Bullion):	0.00
				Viscosity of the silicate slag, Poise:		DrossRecycle:	0.00
						SO2 controller	

Master Metal's 3.6-m SRF Using 100% O₂ to Burner: Simulates 1st Hour After Charge Door is Closed.

Gas firing rate:	Summary of Heat Losses, Kcal/hr During 1st part of 1st charge cycle		Approx. st SO ₂ vented from plant per yr	Estimated ΔT drop due to heat loss when gas passes through the knockout tanks prior to initial dilutn.
Kcal/hr	1,427,348	Loss through furnace walls	96,507	0.42
Btu/hr	5,663,719	Heat to refractory from system	108,086	SCFM to Stack: 17974.
Kg Nat. Gas/hr	107.6	Radiant heat loss thru closed door	4,942	
Lb NG/hr	237.4	Sum of heat losses:	209,535	Gas Temp. exiting stack: 35 degC
% Tonge O ₂ Use	100.0%	Initial T°C of refractory (Ave.)	700 degC	ACFM from the furnace: 5,004
% Exc. O ₂ → NG	-20%	Final T°C of refractory (Ave.)	725 degC	SCFM from the furnace: 1,221
Ave. Offgas T°C	900 degC	Coke Content of Oxide Component:		ACFM to BH: 6,708
P, inch H ₂ O	0.25	Furnace area in use, m ²	57.7	SCFM to BH: 5,079
Shell, T°F	425 degF	Temperature of bag house:	105 degC	Matte/slag concentration ratio for coke: 0.00
Tramp Air in Through the Door, kg/hr:	835.	Pb in Silicate Phase of Emulsion, mt		SCFM moisture to stack: 766
Percent of Dust Exiting Per Hour	16.7%	Pb in Matte phase of Emulsion, %		Coke Calculated via total heat? Indirectly. 700.
Percent of Charge Fused Per Hour	0.0%	Hrs of 1st Cycle	Furnace Offgas Volume % at 900 degC	Portion of excess coke burned, %/hr? 17.8
Percent of Charge Reduced Per Hour	0.0%	CycleDoorClosed	0 ₂ : 2.1% N ₂ : 26.9% H ₂ O: 58.5% CO ₂ : 12.6% CO: 0.0% SO ₂ : 0.0042% sum= 100.00%	Number of Hrs in this period: One 1.0
Stoic. Pure O ₂ for Natural Gas, kg/hr	430.4	Total Hrs Burner		Soda ash CO ₂ still in solids? Yes 100.0%
Stoic. Pure O ₂ for solid C-> CO ₂ , Kg/hr	0.0	% Dross Formed?		Soda ash CO ₂ to gas vent? No 0.0%
Initial Temperature of Charge: 90 degC	0	0		Coke heat input, Kcal/hr
Final Temperature of Charge: 310 degC	1	1		System's heat gain from refractory, Kcal/hr
Charge Moisture Removed:	66.7%			Heat lost to furnace offgas & dust, Kcal/hr
Offgas volume at average furnace temperature:				Heat to furnace charge from burner, Kcal/hr
	8,503 ACMH			Heat lost from walls and door, Kcal/hr
Dilution air to cooler, kg/hr	3,215			Sum of heats of reaction and fusion:
Dilutn. air to cooler, ACMH	5,885			Air drawn into furnace via door, SCFM 382
Total ACMH into cooler	14,388			Furnace Gas, kg " ", kgMole 84.9
Comments: One third of charges H ₂ O evaporated during charging process. Note that exit gas temperature is the average during the period.				Gas Volume to Baghouse, ACMH = 11,398
				v/o CO ₂ diluted 0.86% Sn(Bullion): 0.00
				v/o SO ₂ diluted 0.00029% Sb(Bullion): 0.00
				Cu(Bullion): 0.00
				EFFICIENCY OF HEAT UTILIZATION, %
			Space between door and furnace, inches =	0.39
			Start Up Controller for Totals =	1.00
			SO ₂ , %/hr	60.0%
			Lb SO ₂ /Hr	0.506
			SO ₂ evoln., hrs	1.0
			Press., mm:	730
			Ambient, degC:	7

Master Metal's 3.6-m SRF Using 100% O₂ to Burner: Simulates 1st Hour After Charge Door is Closed.

				Efficiency
Kg moles of gas from furnace	84.9	Oxide Slag mt	5.93	Kcal from solids 598,155
Kg moles air cooler entry	111.5	C with Oxide	0.67	Feed gases -44,410
Kg moles gas into cooler	196.3	%C in Oxide	11.3%	Kcal from H ₂ O evaporation 0
Air to cooler exit, kg/hr	4518.	C Burnt, mt	0.014	Kcal from coke excess -108,579
Air to cooler exit, ACMH	5059.	C in dust, mt	0.00	Kcal from shell loss 204,593
Kg moles air to cooler exit	156.6	Dilution air to stack, kg/hr:	25450.	TOTAL 649,759
Kg moles, all gases, to BH	353.0	Total gas to stack, kgMoles/hr	1249.	Burner Kcal 1,427,348
ACMH to BH, TOTAL	11398.	Total ACFM into stack	19341.	Efficiency 45.5%
	Atomic Weight	Total Moles (inc. refractory)	Total mt	02 released by redn., mt: 0.0000
				SCFH air drawn into furnace: 22,924
				SCFM air drawn into furnace: 382
PbO	223	0.00811	1.809	
Pb*	207	0.04642	9.608	
C*, non-gas	12	0.05703	0.684	
CO ₂	44	0.00190	0.083	
Fe*	55.8	0.00337	0.188	
FeO	71.8	0.00148	0.107	
SO ₃	80	0.00237	0.190	
S	32	0.00360	0.115	
Na(K)2O	62	0.00466	0.289	
SiO ₂	60	0.02006	1.204	
Ca(Ba)O	56	0.00117	0.065	
MgO	40.3	0.00263	0.106	
Cr ₂ O ₃	152	0.00011	0.017	
Al ₂ O ₃	102	0.00076	0.077	
NaCl	58.5	0.00231	0.135	
SnO	135	0.00039	0.052	
Sb ₂ O ₃	291	0.00074	0.216	
H ₂ OFrCb	18	0.03050	0.549	
ZnO	81.4	0.00026	0.021	
SO ₂	64.0	0.00000	0.000	
O ₂	32.0	0.01678	0.537	
CuO	79.5	0.00074	0.059	
Sum:		0.20538	16.112	

Master Metal's 3.6-m SRF Using 100% O₂ to Burner: Simulates 1st Hour After Charge Door is Closed.

Weight %	P. Cake	Plates	MMdust	MMDrs	Coke	Fe Chip	Air	Oxygen	Lenox	NiterDs	Silica	Anzon	NatGas
PbO	35.5%	19.0%	35.3%	0.0%	0.0%	0.0%	0.00%	0.00%	34.9%	10.0%	0.00%	71.1%	0.0%
Pb*	8.0%	73.2%	0.0%	87.3%	0.0%	0.0%	0.00%	0.00%	0.0%	61.2%	0.0%	0.0%	0.00%
C*	12.00%	1.50%	0.00%	2.3%	92.2%	3.00%	0.00%	0.00%	0.0%	0.50%	0.00%	22.1%	75.00%
CO ₂	6.00%	0.00%	0.00%	0.00%	0.00%	0.00%	0.00%	0.00%	0.0%	0.00%	0.00%	0.00%	0.00%
Fe*	0.00%	0.1%	0.00%	0.00%	0.00%	82.50%	0.00%	0.00%	0.0%	0.00%	0.00%	0.0%	0.00%
FeO	2.80%	0.1%	0.50%	0.50%	0.29%	12.00%	0.00%	0.00%	0.9%	0.50%	0.30%	0.0%	0.00%
SO ₃	1.00%	2.0%	12.0%	0.00%	0.00%	0.00%	0.00%	0.00%	0.0%	0.00%	0.00%	4.0%	0.00%
S	0.00%	0.00%	0.00%	1.70%	5.50%	0.00%	0.00%	0.00%	0.3%	1.50%	0.00%	0.00%	0.00%
Na(K)2O	1.10%	0.00%	35.0%	2.00%	0.10%	0.00%	0.00%	0.00%	3.5%	7.30%	0.01%	0.00%	0.00%
SiO ₂	11.00%	1.00%	0.50%	1.50%	1.50%	2.50%	0.00%	0.00%	36.0%	0.50%	99.6%	0.99%	0.00%
Ca(Ba)O	1.50%	0.15%	0.10%	0.60%	0.09%	0.00%	0.00%	0.00%	0.90%	5.00%	0.00%	1.50%	0.00%
MgO	7.50%	0.00%	0.10%	0.02%	0.04%	0.00%	0.00%	0.00%	0.70%	0.10%	0.00%	0.00%	0.00%
Cr2O ₃	1.10%	0.00%	0.13%	0.02%	0.00%	0.00%	0.00%	0.00%	0.00%	0.00%	0.00%	0.00%	0.00%
Al ₂ O ₃	5.00%	0.05%	0.10%	0.05%	0.18%	0.00%	0.00%	0.00%	16.1%	0.00%	0.10%	0.00%	0.00%
NaCl	0.06%	0.52%	10.0%	1.00%	0.10%	0.00%	0.00%	0.00%	0.26%	0.50%	0.00%	0.35%	0.00%
SnO	0.01%	0.10%	2.50%	0.60%	0.00%	0.00%	0.00%	0.00%	0.10%	6.00%	0.00%	0.00%	0.00%
Sb ₂ O ₃	0.02%	2.30%	1.50%	1.20%	0.00%	0.00%	0.00%	0.00%	0.10%	6.50%	0.00%	0.00%	0.00%
Other	0.00%	0.00%	0.00%	0.00%	0.00%	0.00%	0.00%	0.00%	0.00%	0.00%	0.00%	0.00%	0.00%
ZnO	0.02%	0.00%	2.00%	0.20%	0.00%	0.00%	0.00%	0.00%	1.00%	0.20%	0.00%	0.00%	0.00%
SO ₂	0.000%	0.000%	0.000%	0.000%	0.000%	0.000%	0.00%	0.00%	0.000%	0.000%	0.000%	0.000%	0.000%
O ₂	0.000%	0.000%	0.000%	0.000%	0.000%	0.000%	23.20%	100.0%	0.000%	0.000%	0.000%	0.000%	0.000%
CuO	0.01%	0.00%	0.25%	1.00%	0.00%	0.00%	0.00%	0.00%	0.00%	0.20%	0.00%	0.00%	0.00%
H ₂ OXL	7.33%	0.00%	0.00%	0.00%	0.00%	0.00%	0.00%	0.00%	5.20%	0.00%	0.00%	0.00%	0.00%
CO	0.000%	0.000%	0.000%	0.000%	0.000%	0.000%	0.00%	0.00%	0.000%	0.000%	0.000%	0.000%	0.000%
H ₂	0.000%	0.000%	0.000%	0.000%	0.000%	0.000%	0.00%	0.00%	0.000%	0.000%	0.000%	0.000%	25.0%
N ₂	0.000%	0.000%	0.000%	0.000%	0.000%	0.000%	76.8%	0.0%	0.000%	0.000%	0.000%	0.000%	0.000%
Sum:	100.0%	100.0%	100.0%	100.0%	100.0%	100.0%	100.0%	100.0%	100.0%	100.0%	100.0%	100.0%	100.0%
H₂O Fr.	23.3%	1.3%	0.7%	2.0%	2.0%	1.2%	0.5%	0.0%	15.3%	2.0%	1.3%	0.0%	0.0%
P. Cake	Plates	MMdust	MMDrs	Coke	Fe Chip	Air	Oxygen	Lenox	NiterDs	Silica	Anzon	NatGas	

Moisture contents of solid feeds are multiplied by 2/3 to reflect partial drying during the charging process.

Master Metal's 3.6-m SRF Using 100% O₂ to Burner: Simulates 1st Hour After Charge Door is Closed.

Moles	P. Cake	Plates	MMdust	MMDr's	Coke	Fe Chip	Air	Oxygen	Lenox	NiterDs	Silica	Anzor	NatGas
PbO	0.0022	0.0052	0.0007	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000
Pb ⁻	0.0005	0.0215	0.0000	0.0244	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000
C ⁻	0.0139	0.0076	0.0000	0.0111	0.0239	0.0006	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0067
CO ₂	0.0019	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000
Fe ⁻	0.0000	0.0001	0.0000	0.0000	0.0000	0.0033	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000
FeO	0.0005	0.0001	0.0000	0.0004	0.0000	0.0004	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000
SO ₃	0.0002	0.0015	0.0007	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000
S	0.0000	0.0000	0.0000	0.0031	0.0005	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000
Na(K)20	0.0002	0.0000	0.0025	0.0019	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000
SiO ₂	0.0025	0.0010	0.0000	0.0014	0.0001	0.0001	0.0000	0.0000	0.0000	0.0000	0.0000	0.0148	0.0000
Ca(Ba)0	0.0004	0.0002	0.0000	0.0006	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000
MgO	0.0026	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000
Cr203	0.0001	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000
Al203	0.0007	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000
NaCl	0.0000	0.0005	0.0008	0.0010	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000
SnO	0.0000	0.0000	0.0001	0.0003	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000
Sb203	0.0000	0.0005	0.0000	0.0002	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000
H20FrCb	0.0180	0.0045	0.0002	0.0064	0.0003	0.0001	0.0002	0.0000	0.0000	0.0000	0.0007	0.0000	0.0000
ZnO	0.0000	0.0000	0.0001	0.0001	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000
SO ₂	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000
O ₂	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0060	0.0108	0.0000	0.0000	0.0000	0.0000	0.0000
CuO	0.0000	0.0000	0.0000	0.0007	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000
H20XL	0.0057	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000
CO	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000
H ₂	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0135	0.0000
N ₂	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0228	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000
Sum:													
H20 Fr.	P. Cake	Plates	MMdust	MMDr's	Coke	Fe Chip	Air	Oxygen	Lenox	NiterDs	Silica	Anzor	NatGas

Master Metal's 3.6-m SRF Using 100% O₂ to Burner: Simulates 1st Hour After Charge Door is Closed.

ESTIMATE OF STACK EMISSIONS (FROM NEW BAGHOUSE) WHEN OPERATING THE 3.6-M ROTARY FURNACE,

and Summary of Results of Furnace and Flue Calculations

	Drying		Melting		Reduction
Operation time during each furnace charging cycle:	1.0	Hr		Hr	Hr
Initial temperature for drying, melting, or reduction:	194	°F		°F	°F
Temperature at finish of drying, melting, or reduction:	590	°F		°F	°F
Natural gas firing rate:	5.7 MM	Btu/Hr		Btu/Hr	Btu/Hr
Average off-gas temperature from the furnace:	1652	°F		°F	°F
Natural gas consumed by the furnace:	5260	Ft ³ /Hr		Ft ³ /Hr	Ft ³ /Hr
Oxygen consumed by the furnace:	8978	Ft ³ /Hr		Ft ³ /Hr	Ft ³ /Hr
Diluted flue gas to the stack					
Flow rate entering stack @ 95 °F :	19,341	ACFM		ACFM	ACFM
Flow, DSCFM = 77°F @ 1 atm.:	17,208	DSCFM		DSCFM	DSCFM
Moisture content of gas to stack:	4.3%	Vol%		Vol%	Vol%
Particulates to stack, grains/dscf:	0.0026	g/DSCF		g/DSCF	g/DSCF
SO ₂ to the stack:	0.51	Lb/Hr		Lb/Hr	Lb/Hr
Emissions from Stack (Lb/Hr)					
Tetraethyl lead:	*BDL	Lb/Hr	*BDL	Lb/Hr	*BDL
Total chlorine expressed as chloride:	0.023	Lb/Hr		Lb/Hr	Lb/Hr
Lead as Pb:	0.13	Lb/Hr		Lb/Hr	Lb/Hr
Antimony as Sb:	0.005	Lb/Hr		Lb/Hr	Lb/Hr
Arsenic as As:	**ID	Lb/Hr	**ID	Lb/Hr	**ID
Barium as Ba:	<0.01	Lb/Hr		Lb/Hr	Lb/Hr
Silver as Ag:	<0.01	Lb/Hr		Lb/Hr	Lb/Hr
Cadmium as Cd:	<0.01	Lb/Hr		Lb/Hr	Lb/Hr
Chromium as Cr	0.0003	Lb/Hr		Lb/Hr	Lb/Hr
Mercury, thallium, and beryllium:	***NIF		***NIF		***NIF
* Below Detectable Limit	** Insufficient data to estimate.		*** Not known to be in feedstock to plant.		

Master Metal's 3.6-m SRF Using 100% O₂ to Burner: Simulates 1st Hour After Charge Door is Closed.

mt	Bullion	Mat+Slag	Dust Out	mt	All mt In	All mt Out	MoleChecks	At. Wt.		PbO
PbO	0.00000	1.80894	0.16011	PbO	1.80894	1.96905	0.054529	223		PbO
Pb*	9.45964	0.00000	0.00000	Pb*	9.60826	9.45964	0.054529	207		Pb*
C*	0.00000	0.65924	0.00000	C*	0.76502	0.65924	0.967361	12		C*
CO2	0.00000	0.00000	0.65849	CO2	0.74190	1.12979	0.967361	44		CO2
Fe*	0.00000	0.00000	0.00000	Fe*	0.18779	0.00000	0.00485	55.8		Fe*
FeO	0.00000	0.34598	0.00224	FeO	0.10659	0.34823	0.00485	71.8		FeO
SO3	0.00000	0.00000	0.05375	SO3	0.18959	0.05375	0.005972	80		SO3
S	0.00000	0.16950	0.00000	S	0.11528	0.16950	0.005972	32		S
Na2O	0.00000	1.05978	0.15689	Na(K)2O	1.21667	1.21667	RefrConsum	62		Na(K)2O
SiO2	0.00000	1.20151	0.00224	SiO2	1.20375	1.20375	0.00000	60		SiO2
CaO	0.00000	0.06491	0.00045	Ca(Ba)O	0.06536	0.06536	0.00000	56		Ca(Ba)O
MgO	0.00000	0.10554	0.00045	MgO	0.10599	0.10599	0.00000	40.3		MgO
Cr2O3	0.00000	0.01645	0.00058	Cr2O3	0.01703	0.01703	0.00000	152		Cr2O3
Al2O3	0.00000	0.07689	0.00045	Al2O3	0.07734	0.07734	0.00000	102		Al2O3
NaCl	0.00000	0.09858	0.04483	NaCl	0.14340	0.14340	0.00000	58.5		NaCl
SnO	0.00000	0.04092	0.01121	SnO	0.05213	0.05213	Sulfide	135		SnO
Sb2O3	0.00000	0.20952	0.00672	Sb2O3	0.21624	0.21624	339.0	291		Sb2O3
H2OFrCb	0.00000	0.00000	0.00000	H2OFrCb	0.54897	0.79107	Atomic	18		H2OFrCb
ZnO	0.00000	0.01187	0.00897	ZnO	0.02083	0.02083	97.4	81.4		ZnO
SO2	0.00000	0.00000	0.00023	SO2	0.00000	0.00023	Weights	64		SO2
O2	0.00000	0.00000	0.00006	O2	0.53707	0.05582	86.9	32		O2
CuO	0.00000	0.05790	0.00112	CuO	0.05902	0.05902	159.0	79.5		CuO
H2OXL	0.00000	0.00000	0.00000	H2OXL	0.10194	0.10194		18		H2OXL
CO	0.00000	0.00000	0.00000	CO	0.00000	0.00000		28.00		CO
H2	0.00000	0.00000	0.00000	H2	0.02690	0.00000		2.00		H2
N2	0.00000	0.00000	0.00000	N2	0.63804	0.63804		28.00		N2
Sum:	9.4596	5.9275	1.108778	Sum:	18.55406	18.55407	CHECK			
	Bullion	Mat+Slag	Dust Out							
			1.108778							
			Check							
							O2Mole Σ			
							1.92252			
							1.92252			

Master Metal's 3.6-m SRF Using 100% O₂ to Burner: Simulates 1st Hour After Charge Door is Closed.

Moles To Flux	
0.0074	PbO
0.0464	Pb*
0.0638	C*
0.0169	CO ₂
0.0034	Fe*
0.0015	FeO
0.0017	SO ₃
0.0036	S
0.0171	Na(K)2O
0.0200	SiO ₂
0.0012	Ca(Ba)O
0.0026	MgO
0.0001	Cr ₂ O ₃
0.0008	Al ₂ O ₃
0.0017	NaCl
0.0003	SnO
0.0007	Sb ₂ O ₃
0.0439	H ₂ OFrCb
0.0001	ZnO
0.0000	SO ₂
0.0168	O ₂
0.0007	CuO
0.0057	H ₂ OXL
0.0000	CO
0.0135	H ₂
0.0228	N ₂
Moles To Flux	

Master Metal's 3.6-m SRF Using 100% O₂ to Burner: Simulates 1st Hour After Charge Door is Closed.

	mt	Total In	Total Out	mt	At. Wt.	m3/tmole	
PbO	1.8089	1.9691		PbMoleSum	223	25.93	PbO
Pb*	9.6083	9.4596		11.2874	207	32.34	Pb*
C*	0.7650	0.6592		11.2874	12	5.33	C*
CO ₂	0.7419	1.1298		Fe&S m.s.	44	0.00	CO ₂
Fe*	0.1878	0.0000		0.00485	55.8	14.68	Fe*
FeO	0.1066	0.3482		0.00485	71.8	15.28	FeO
SO ₃	0.1896	0.0537		0.00597	80		SO ₃
S	0.1153	0.1695		0.00597	32	16.00	S
Na(K)2O	1.2167	1.2167		Refractory	62	31.00	Na(K)2O
SiO ₂	1.2037	1.2037		0.00000	60	27.27	SiO ₂
Ca(Ba)O	0.0654	0.0654		0.00000	56	18.67	Ca(Ba)O
MgO	0.1060	0.1060		0.00000	40.3	12.21	MgO
Cr2O ₃	0.0170	0.0170		0.00000	152	34.55	Cr2O ₃
Al2O ₃	0.0773	0.0773		0.00000	102	31.88	Al2O ₃
NaCl	0.1434	0.1434		0.00000	58.5	34.41	NaCl
SnO	0.0521	0.0521		C moleSum	135	20.93	SnO
Sb2O ₃	0.2162	0.2162		0.08061	291	55.96	Sb2O ₃
H ₂ OFrCb	0.5490	0.7911		0.08061	18	0.00	H ₂ OFrCb
ZnO	0.0208	0.0208		H wt Sum	81.4	14.88	ZnO
SO ₂	0.0000	0.0002		0.09922	64		SO ₂
O ₂	0.5371	0.0558		0.09922	32		O ₂
CuO	0.0590	0.0590		O wt Sum	79.5	12.42	CuO
H ₂ OXL	0.1019	0.1019		1.92252	18		H ₂ OXL
CO	0.0000	0.0000		1.92252	28.0		CO
H ₂	0.0269	0.0000		Check?	2.0		H ₂
N ₂	0.6380	0.6380			28.0		N ₂
Sum:	18.5541	18.5541			At. Wt.	m3/tmole	
	mt In	mt Out					
							g/cc-ft ³ /lb 62.42

Master Metal's 3.6-m SRF Using 100% O₂ to Burner: Simulates 1st Hour After Charge Door is Closed.

Boundary Conditions for Operating the Short Rotary Furnace		H fusion kcal/kgmole	Fusion Kcal/hr
Dust recycle rate, dry mt/hr	0.45	PbO	6570
Calculated Natl. Gas Kcal/hr Input	1,427,349	Pb°	1150
Calculated Natl. Gas used, mt/hr	0.108	C	
Calculated Natl. Gas used, kg/hr	107.60	CO ₂	
Natl. Gas heat content (net), kcal/kg	13,265	FeS	7730
Natl. Gas density, lb/ft ³ @ STP	0.044	FeO	7500
Fuel heat capacity, kgcal/kg°C	0.55	SO ₃	
Natl. Gas density, kg/m ³ @ STP	1.44	Na(K)2O	11400
Fuel temperature, C	7	SiO ₂	2600
Air temperatures, C	7	Ca(Ba)O	12200
Solids feed conc. temperature, C	7	MgO	18500
Return dust temperature, C	7	Cr ₂ O ₃	4000
Offgas temperature, C	310	Al ₂ O ₃	25700
Final solids temperature, °C	310	NaCl	6700
Stoic. oxygen for fuel, kg dry/kg fuel	4.00	SnO	6000
% excess oxygen for natural gas	-0.20	Sb ₂ O ₃	13300
Air inleakage to furnace, mt/hr	0.83	CuO	13580
Atomizing steam, kcal/kg	0.00	ZnO	4500
Atomizing steam, kg/kg fuel	0.00	Sum =	141430.
Sum H ₂ O in furnace solids, mt/hr	0.65		370,846
Moisture in air&O ₂ &Gas, mt /hr	0.00	Chemical Reactions	
Total moisture to plant, mt/hr	0.65	PbO + 1/2C = Pb° + 1/2CO ₂	24.5
Furnace shell temperature, F	425	PbS + Na ₂ O = Na ₂ S + 1/2O ₂	139.3
Furnace (small) shell area, m ²	57.65	Fe° + PbS = FeS + Pb°	21.3
Heat loss, kiln shell & door, kcal/hr	96,507	PbS + FeO = FeS + PbO	145.7
		PbO ₃ S + Na ₂ O = Na ₂ S + PbO + 2O ₂	2276.9
		Na ₂ CO ₃ = Na ₂ O + CO ₂	725.0
		Large=	69.08

Master Metal's 3.6-m SRF Using 100% O₂ to Burner: Simulates 1st Hour After Charge Door is Closed.

DrssProdmt	Dross, %	Normalized Silicate Fraction i.e., Slag Assay	PbO;PbS
0.00000	8.000%		Pb*
0.00000	86.165%		C*
0.00000	0.000%		CO ₂
0.00000	0.000%		Fe*;FeS
0.00000	0.000%	3.5% Al ₂ O ₃ +Cr ₂ O ₃	FeO
0.00000	0.000%	10.9% FeO	SO ₃
0.00000	0.000%	-	S;Na ₂ S
0.00000	5.000%	-	Na(K)2O
0.00000	0.000%	40.4% Na ₂ O	SiO ₂
0.00000	0.000%	45.1% SiO ₂	Ca(Ba)O
0.00000	0.000%	100.0% TOTAL	MgO
0.00000	0.000%		Cr ₂ O ₃
0.00000	0.000%	Heat S In as S: 0.1911	Al ₂ O ₃
0.00000	0.000%	Heat S Out as S 0.1911	NaCl
0.00000	0.133%	Heat SO ₂ as S: 0.000191	SnO
0.00000	0.551%	SDross(Heat) 0.0000	Sb ₂ O ₃
0.00000	0.000%	Ssilicate(Heat) 0.1164	H ₂ OFrCb
0.00000	0.000%	Smatte(Heat) 0.0532	ZnO
0.00000	0.000%	S in dust=Heat 0.0217	SO ₂
0.00000	0.000%	S out by addn. 0.1914	O ₂
0.00000	0.151%		CuO
0.00000	0.000%		H ₂ OXL
0.00000	0.000%	FlueGasMt/hr 2.0584	CO
0.00000	0.000%		H ₂
0.00000	0.000%	Wt. % SO ₂ 0.0000%	N ₂
0.00000	100.0%	SO ₂ exit/hr, Short Tons this furnace: 0.000253	SUM 0
		SO ₂ exit/yr, st/yr all furnaces: 0.42	
		% of S as Silicate 60.9%	Total % = Check
		% of S as Matte 27.8%	
		% of S as Dross	
		% of S as Dust 11.3%	
		% of S as Gas 0.10%	100%

Master Metal's 3.6-m SRF Using 100% O₂ to Burner: Simulates 1st Hour After Charge Door is Closed.

mt of CO ₂ =	0.658	
O from PbO=	0.000	but not dust.
O for Fe=	0.000	but not dust.
O from SO ₃ =	0.000	but not dust.
O from SnO=	0.0000	but not dust.
O from Sb ₂ O ₃ =	0.0000	but not dust.
O from ZnO=	0.0000	but not dust.
O from CuO=	0.0000	but not dust.
O for Si in Fe°	0.0000	
mt eq CO ₂ in=	0.6585	
		FeOMoleFract.
		0.07
		CO ₂ /COwt.=R 1000.00
		1.27R + 1 = 1271.0000

$$(CO_2-x)/1.27x = CO_2/CO \text{ wt=R}$$

$$(CO_2-x) = 1.27Rx$$

$$CO_2 \text{ decomposed} = 1.27Rx + x = x(1.27R+1)$$

x= 0.001 mt CO₂ decomposed

1.27x= 0.001 mt CO evolved from melt.

CO₂-x= 0.658 mt CO₂ evolved from melt.

0.000 mt Carbon needed for above.

0.398 mt C

0.659 mt carbon for slag; (inc. H₂)

0.37

mt

20. % Excess Coke

Master Metal's 3.6-m SRF Using 100% O₂ to Burner: Simulates 1st Hour After Charge Door is Closed.

Density 11.30	PbO	mt O ₂ Out	m ³ Matte/Slag	mt O ₂ In	PbO	
Pb°	Pb°	0.14128	0.21034	0.12979	Pb°	
C°	C°	0.00000	0.00000	0.00000	C°	
CO ₂	CO ₂	0.82166	0.00000	0.53956	CO ₂	
Fe°	Fe°	0.00000	0.00000	0.00000	Fe°	m ³ of Bullion: 0.84
FeO	FeO	0.07760	0.07361	0.02375	FeO	m ³ of Slag: 1.93
SO ₃	SO ₃	0.03225	0.00000	0.11376	SO ₃	m ³ of Dross: 0.00
S	S	0.00000	0.08475	0.00000	S	m ³ in Charge: 2.77
Na(K)2O	Na(K)2O	0.31398	0.52989	0.31398	Na(K)2O	FreshBrckM3 6.9
SiO ₂	SiO ₂	0.64200	0.54614	0.64200	SiO ₂	WornBrickM3 8.8
Ca(Ba)O	Ca(Ba)O	0.01867	0.02164	0.01867	Ca(Ba)O	Effective Furn. Volume with New Brick vs with Worn Brick.
MgO	MgO	0.04208	0.03198	0.04208	MgO	
Cr ₂ O ₃	Cr ₂ O ₃	0.00538	0.00374	0.00538	Cr ₂ O ₃	
Al ₂ O ₃	Al ₂ O ₃	0.03640	0.02403	0.03640	Al ₂ O ₃	
Sn°	NaCl	0.00000	0.05799	0.00000	NaCl	
6.60	SnO	0.00618	0.00634	0.00618	SnO	
6.40	Sb ₂ O ₃	0.03567	0.04029	0.03567	Sb ₂ O ₃	
Sb°	H ₂ OFrCb	0.21521	0.00000	0.00000	H ₂ OFrCb	
ZnO	ZnO	0.00410	0.00217	0.00410	ZnO	
SO ₂	SO ₂	0.00011	0.00000	0.00000	SO ₂	
O ₂	O ₂	0.05582	0.00000	0.53707	O ₂	
CuO	CuO	0.01188	0.00905	0.01188	CuO	
H ₂ OXL	H ₂ OXL	0.09062	0.00000	0.09062	H ₂ OXL	
CO	CO	0.00000	0.00000	0.00000	CO	
H ₂	H ₂	0.00000	0.00000	0.00000	H ₂	
N ₂	N ₂	0.00000	0.00000	0.00000	N ₂	
Sum =	Oxygen Out =	2.55088	1.93496	2.55088	<- Oxygen In	
			SUM OF M ³	Oxygen In		

Master Metal's 3.6-m SRF Using 100% O₂ to Burner: Simulates 1st Hour After Charge Door is Closed.

Heat Capacity Calc., kcal/kg			Mass Basis mt/hr	Heat Inputs kcal/hr	Mass Basis mt/hr	Heat Outputs kcal/hr
T1, C= 90	for solids		Fuel Combustion	1,427,349	Sensible Heats:	
T2, C= 310	for solids			0.108	C0 ₂ exhaust dry gas	108,019
Average Feed 0.180	4.3235E-05		Steam for Atomization	0	0.47	
C0 ₂ 229	kcal/kg		Coke Combustion:	108,579	C0 exhaust dry gas	0
C0 234				0.0141	02 exhaust dry gas	11,877
Dust recy. 0.184	0.0000838		Sensible Heats:		0.06	
T1, C= 25.	for gases		Dry furnace feedstocks	190,289	N2 exhaust dry gas	148,695
T2, C= 900.	for gases			16.26	0.64	
O2 213	kcal/kg		Dry dust recycle	5,385	S0 ₂ exhaust dry gas	37
S0 ₂ 163				0.45	0.00023	
Air 0.234	0.0000173		Natural gas	417	Solids in Charge	891,459
H2O vapor 443		Sensible Heat		0.108	(heated/hr) 16.26	
N2 233	kcal/kg		Gas combustion oxygen	564	Dry dust in offgas	10,949
H2O Vaporize 598	Heat of boiling			0.34	0.18	
			Total air into furnace	1,361	Slag discharge is zero	0
				0.83	0.00	
			H2O in all input gases	29	H2O from all input gases	1,839
				0.004	0.004	
			Free H2O in solid feeds	42,039	Free H2O in solid feeds	286,393
				0.65	0.65	
			FeS-Fe°&FeO Formation ΔH	0		
			Na2Sfrom PbS	"	H2O from combustion	107,208
			Na2Sfrom S0 ₃	"	0.24	
			Na2O-Na2C0 ₃	"	H2O from atomization steam	0
			Pb° from PbO	"	0.00	
			Refractory dissolved:	0	H2O from solid Rx product:	0
				0.0000	0.00	
			Heats of fusion and vaporization:			
			Fusion	0		
			Vaporization=already in ΔH	0	Shell+Door+Refr heat losses	209,535
			Total Heat Input, kcal/hr	1,776,012	Total Heat Out, kcal/hr:	1,776,012

Master Metal's 3.6-m SRF Using 100% O₂ to Burner: Simulates 1st Hour After Charge Door is Closed.

	Slag Wt,mt	Matte Wt,mt		
PbO	1.80894	0.00000	-	11.3% % fixed C in matte + slag. 0.0% is PbS in Matte
-	0.00000	0.00000	PbS	is Matte + Slag factor kg Pb loss/mt bullion.
C	0.65924	0.00000	C	0.00 is (matte/slag) concentration ratio for coke
CO ₂	0.00000	0.00000	CO ₂	5.0% is the % Na ₂ S in slag
-	0.00000	0.06007	FeS	2.03% is the percent sulfur in slag.
FeO	0.29085	0.00601	FeO	
	0.00000	0.00000	-	
Na ₂ S	0.28361	0.05337	Na ₂ S	
Na(K)2O	0.79192	0.00000	-	
SiO ₂	1.20133	0.00018	SiO ₂	0.005 is (matte/slag) concentration ratio for silica
Ca(Ba)O	0.06490	0.00001	Ca(Ba)O	0.005 is (matte/slag) concentration ratio for lime
MgO	0.10553	0.00002	MgO	0.005 is (matte/slag) concentration ratio for magnesia
Cr ₂ O ₃	0.01645	0.00000	Cr ₂ O ₃	0.005 is (matte/slag) concentration ratio for chromia
Al ₂ O ₃	0.07688	0.00001	Al ₂ O ₃	0.005 is (matte/slag) concentration ratio for alumina
NaCl	0.09809	0.00049	NaCl	0.17 is (matte/slag) concentration ratio for chloride
SnO	0.04033	0.00059	CdO	0.50 is (matte/slag) concentration ratio for tin
Sb ₂ O ₃	0.20787	0.00192	Sb ₂ S ₃	0.27 is (matte/slag) concentration ratio for antimony
H ₂ OFrCb	0.00000	0.00000	H ₂ OFrCb	
ZnO	0.01153	0.00040	ZnS	1.00 is (matte/slag) concentration ratio for Zn
SO ₂	0.00000	0.00000	SO ₂	
O ₂	0.00000	0.00000	O ₂	
CuO	0.01473	0.04317	Cu ₂ S	100.00 is (matte/slag) concentration ratio for Cu
-	0.00000	0.00000	-	
-	0.00000	0.00000	-	
-	0.00000	0.00000	-	
-	0.00000	0.00000	-	
TOTAL SLAG	5.67221	0.16623	TOTAL Matte	
	Slag Wt,mt	Matte Wt,mt		

Master Metal's 3.6-m SRF Using 100% O₂ to Burner: Simulates 1st Hour After Charge Door is Closed.

Calculation of Slag Viscosity					
		Wt %	Weight	MoleFrac.Slag	MoleNonMatte
X' SiO ₂ ' =	Moles Si+P+Ti	PbO	37.3%	1.81	0.1442
X' CaO' =	Moles Ca+Mg+Mn+Pb+Zn+Fe	Pb ⁺	0.0%	0.00	0.0000
X' Al ₂ O ₃ ' =	Moles Al ₂ O ₃ +Cr ₂ O ₃	C ⁺	0.0%	0.00	0.0000
X' Na ₂ O' =	Moles NaK ₂ O+Na ₂ S (non-mat)	CO ₂	0.0%	0.00	0.0000
X' CaF ₂ ' =	Moles NaCl	Fe ⁺	0.0%	0.00	0.0000
A=		FeO	6.0%	0.29	0.0720
B=		SO ₃	0.0%	0.00	0.0000
		Na ₂ S	2.4%	0.12	0.0646
		Na(K)2O	16.3%	0.79	0.2271
		SiO ₂	24.8%	1.20	0.3559
		Ca(Ba)O	1.3%	0.06	0.0206
		MgO	2.2%	0.11	0.0466
		Cr ₂ O ₃	0.3%	0.02	0.0019
		Al ₂ O ₃	1.6%	0.08	0.0134
		NaCl	2.0%	0.10	0.0298
		SnO	0.8%	0.04	0.0053
		Sb ₂ O ₃	4.3%	0.21	0.0127
		Other	0.0%	0.00	0.0000
		ZnO	0.2%	0.01	0.0025
		SO ₂	0.0%	0.00	0.0000
		O ₂	0.0%	0.00	0.0000
		CuO	0.3%	0.01	0.0033
		H ₂ OXL	0.0%	0.00	0.0000
		CO	0.0%	0.00	0.0000
		H ₂	0.0%	0.00	0.0000
		N ₂	0.0%	0.00	0.0000
		Sum:	100.00%	4.85	1.0000
					0.05625

Master Metal's 3.6-m SRF Using 100% O₂ to Burner: Simulates 1st Hour After Charge Door is Closed.

			100	Start-up controller		
Calculation of the low-temperature heat capacities						
	A	T ₂ - T ₁	B	T _{2sq} - T _{1sq}	C	1/T ₂ - 1/T ₁
CO ₂	10.57	25	1.05E-03	1.43E+04	2.06E+05	-3.07E-04
CO	6.708	25	5.53E-04	1.43E+04	6.20E+03	-3.07E-04
Sb406 vap.	52.016	25	1.69E-03	1.43E+04	8.30E+05	-3.07E-04
O ₂	7.16	25	5.00E-04	1.43E+04	4.00E+04	-3.07E-04
SO ₂	11.04	25	9.40E-04	1.43E+04	1.84E+05	-3.07E-04
H ₂ O vapor	7.3	25	1.23E-03	1.43E+04	0.00E+00	-3.07E-04
N ₂	6.83	25	4.50E-04	1.43E+04	1.20E+04	-3.07E-04

Master Metal's 3.6-m SRF Using 100% O₂ to Burner: Simulates 1st Hour After Charge Door is Closed.

Calculation of high-temperature heat capacities, kcal/kg							Low-Temp. Heat Capacities:	
	A	T ₂ - T ₁	B	T _{2sq} - T _{1sq}	C	1/T ₂ - 1/T ₁	T ₁ , °C =	0
							T ₂ , °C =	25
CO ₂	10.57	875	1.05E-03	1.29E+06	2.06E+05	-2.5E-03	CO ₂	4.91E+00
CO	6.708	875	5.53E-04	1.29E+06	6.20E+03	-2.5E-03	CO	6.20E+00
Σ mass for input heat calc:								
Σ mass for output heat calc								
Σ mass in from material bal:								
Sb406 vapor	52.016	875	1.69E-03	1.29E+06	8.30E+05	-2.5E-03	Sb203 vapor	1.83E+00
O ₂	7.16	875	5.00E-04	1.29E+06	4.00E+04	-2.5E-03	O ₂	5.43E+00
SO ₂	11.04	875	9.40E-04	1.29E+06	1.84E+05	-2.5E-03	SO ₂	3.64E+00
H ₂ O vapor	7.3	875	1.23E-03	1.29E+06	0.00E+00	-2.5E-03	H ₂ O vapor	1.11E+01
N ₂	6.83	875	4.50E-04	1.29E+06	1.20E+04	-2.50E-03	N ₂	6.20E+00

Master Metal's 3.6-m SRF Using 100% O2 to Burner: Simulates 2nd and 3rd Hr After Charge Door is Closed.

Furnace:	Old (3.6-m)	Pb ^o in Bulln, mt		CHARGE CYCLE IS FIRST	Bullion	Silicate Only	Matte Only
Date:	23-Jul-91	Pb in Feed, mt	11.19	*****	PbO		
EndOfPeriod T ^c	785 degC	Pb,TotalSlag, mt		*****	Pb ^o , PbS		
Parameters to Check		Pb in Dust, mt		C*			
Slag FeO =		Pb,DrossFloat,mt		CO2			
Target =	10 to 15%	Bullion, mt		FeS			
Basicity =		PbRecToBulln		FeO			
Target =	1.1 to 1.4	Balanced if small	9.47E-06	S			
Viscosity =		SilicatePhase, mt		Na2S			
Target =	10 to 40	MattePhase, mt		Na(K)2O			
Slag Ratio of		Soda/hr to flue	0.045	SiO2			
(Si+Al)/Na =	1.2	Dust, % of feed	3.0%	Ca(Ba)O			
Slag Ratio of		Molds Output-->		MgO			
(Si+Al)/Na =	1.2	DryFeed,lb	DryFeedMt/hr	Cr203			
Feed(Scoops)	Total Slag, %			Al203			
P,CakeTEL(2)	2600.	1.18		NaCl			
MM Plate(6)	13328.	6.04		Sn ⁺ , SnO			
Dust In (2)	990.	0.45		Sb ⁺ ,Sb2S3			
MM Dross(6)	12610.	5.72		Other			
Coke(1/2)	679.	0.31		ZnO,ZnS			
FeChips(0.35)	491.	0.22		Other			
Total Air In	2078.	0.94		Cu,Cu2S			
O2 to Burner	757.	0.34		TOTAL	100.0%	100.0%	100.0%
Lenox(0)	0.	0.00					
NiterDross(0)	0.	0.00					
Glass Sand(2)	1960.	0.89					
Anzon(0)	0.	0.0					
Natural Gas	236.7	0.107					
TOTAL	35730.	16.19					
Flux Summary							
Soda(2+BH)	3500.	1.586		MgCr Brick, kg/hr per mt Bullion	0.0	% of S fixed:	99.90%
Sand (1.5)	1960.	0.888		Al203 Brick, Kg/hr per mt Bullion	0.0	Coke in matte + slag, %	"Slag" Basicity
FeChip(.25)	491.	0.223		Mat.FeS/Na2S	0.50	Distribution Factors	
Coke (0.5)	679.	0.308		Mat.FeO Contr.	10	Sn(Bullion):	0.00
Comments: Lead metallics are melted over a 1 hr period. Oxide reduction is nil.				% Excess Coke	20.0%	Sb(Bullion):	0.00
				PbTo"slag"mt		Cu(Bullion):	0.00
				Viscosity of the silicate slag, Poise:		DrossRecycle:	0.00
						SO2 controller	

Master Metal's 3.6-m SRF Using 100% O₂ to Burner: Simulates 2nd and 3rd Hr After Charge Door is Closed.

Gas firing rate:		Summary of Heat Losses, Kcal/hr During 2nd part of 1st charge cycle		Approx. st SO ₂ vented from plant per yr	Estimated ΔT drop due to heat loss when gas passes through the knockout tanks prior to initial dilutn.
Kcal/hr	1,423,334	Loss through furnace walls	96,507	0.42	127 degC
Btu/hr	5,647,790	Heat to refractory from system	21,617	SCFM to Stack:	11977,
Kg Nat. Gas/hr	107.3	Radiant heat loss thru closed door	6,380		
Lb NG/hr	236.7	Sum of heat losses:	124,504		
% TongeO ₂ Use	100.0%	Initial T°C of refractory (Ave.)	725 degC	Gas Temp. exiting stack:	35 degC
% Exc.O ₂ ->NG	-20%	Final T°C of refractory (Ave.)	735 degC	Gas Temp. exiting convective cooler:	345 degC
Ave. Offgas T°C	980 degC	Coke Content of Oxide Component:		Gas Temp. exiting convective cooler:	195 degC
P, inch H ₂ O	0.25	Furnace area in use, m ²	57.7	ACFM to BH:	4,470
Shell, T°F	425 degF	Temperature of bag house:	105 degC	SCFM to BH:	3,385
Tramp Air in Through the Door, kg/hr:	942.	PbInSilicatePhaseOfEmulsion, mt:		Matte/slag concentration ratio for coke:	0.00
Percent of Dust Exiting Per Hour	16.7%	Pb in Matte phase of Emulsion, %:		SCFM moisture to stack:	232
Percent of Charge Fused Per Hour	31.8	Hrs of 1st Cycle		Coke Calculated via total heat?	Indirectly.
		CycleDoorClosed		Portion of excess coke burned, %/hr?	17.8
Percent of Charge Reduced Per Hour	0.0%	Total Hrs Burner	15.0		
Stoic. Pure O ₂ for Natural Gas, kg/hr	429.2	% Dross Formed?	0	Number of Hrs in this period:	2.0
Stoic. Pure O ₂ for solid C-> CO ₂ , Kg/hr	35.0			Soda ash CO ₂ still in solids?	Yes 100.0%
				Soda ash CO ₂ to gas vent?	No 0.0%
Initial Temperature of Charge:	310 degC	Furnace Offgas Volume% at 980 degC		Coke heat input, Kcal/hr	
Final Temperature of Charge:	785 degC	02: 2.1%		System's heat gain from refractory, Kcal/hr	
Charge Moisture Removed:	0.0%	N2: 49.2%		Heat lost to furnace offgas & dust, Kcal/hr	
Offgas volume at average furnace temperature:	5,595 ACMH	H2O: 26.2%		Heat to furnace charge from burner, Kcal/hr	
Dilution air to cooler, kg/hr	2,266	CO ₂ : 22.6%		Heat lost from walls and door, Kcal/hr	
Dilutn. air to cooler, ACMH	4,148	CO: 0.0%		Sum of heats of reaction and fusion:	
Total ACMH into cooler	9,742	SO ₂ : 0.00169%		Air drawn into furnace via door, SCFM	431
Comments:	Note that the exit gas temperature is the average during the period.		sum= 100.00%	FurnaceGas,kg " ",kgMole	Gas Volume to Baghouse, ACMH = 7,595
				v/oCO ₂ diluted	Sn(Bullion): 0.00
				From Stack	Sb(Bullion): 0.00
				v/oSO ₂ diluted	Cu(Bullion): 0.00
			sum= 1.5201	EFFICIENCY OF HEAT UTILIZATION, %	
				Space between door and furnace, inches =	0.44
				Start up controller for totals =	100
				Calculated Eff.	46%
				Target effic.	47.5%
				BH, ACFM	4,470
				Lb SO ₂ /Hr	SO ₂ evoln.,hrs
				0.125	2.0
				Press., mm:	730
				Ambient, degC:	7

Master Metal's 3.6-m SRF Using 100% O₂ to Burner: Simulates 2nd and 3rd Hr After Charge Door is Closed.

				Efficiency
Kg moles of gas from furnace	52.3	Oxide Slag mt	5.86	Kcal to solids 802,563
Kg moles air cooler entry	78.6	C with Oxide	0.63	Kcal to redn. & fusion 55,134
Kg moles gas into cooler	130.8	%C in Oxide	10.7%	Wall & door losses -102,887
Air to cooler exit, kg/hr	3010.	C Burnt, mt	0.027	Feed gases -2,514
Air to cooler exit, ACMH	3371.	C in dust, mt	0.00	Coke heat input, Kcal/hr -100,980
Kg moles air to cooler exit	104.4	Dilution air to stack, kg/hr:	16958.	TOTAL 651,316
Kg moles, all gases, to BH	235.2	Total gas to stack, kgMoles/hr	832.	Burner Kcal 1,423,334
ACMH to BH, TOTAL	7595.	Total ACFM into stack	12888.	Efficiency 45.8%
	Atomic Weight	Total Moles (inc. refractory)	Total mt	
				O ₂ released by redn., mt: 0.0000
				SCFH air drawn into furnace: 25,863
				SCFM air drawn into furnace: 431
PbO	223	0.00812	1.812	
Pb"	207	0.04594	9.510	
C*, non-gas	12	0.05450	0.654	
CO ₂	44	0.00161	0.071	
Fe*	55.8	0.00334	0.187	
FeO	71.8	0.00139	0.100	
SO ₃	80	0.00233	0.186	
S	32	0.00357	0.114	
Na(K)2O	62	0.00459	0.285	
SiO ₂	60	0.01955	1.173	
Ca(Ba)O	56	0.00110	0.062	
MgO	40.3	0.00224	0.090	
Cr ₂ O ₃	152	0.00010	0.015	
Al ₂ O ₃	102	0.00065	0.067	
NaCl	58.5	0.00229	0.134	
SnO	135	0.00038	0.052	
Sb ₂ O ₃	291	0.00074	0.215	
H ₂ OFrCb	18	0.00026	0.005	
ZnO	81.4	0.00025	0.021	
SO ₂	64.0	0.00000	0.000	
O ₂	32.0	0.01753	0.561	
CuO	79.5	0.00073	0.058	
Sum:		0.17123	15.370	

Master Metal's 3.6-m SRF Using 100% O₂ to Burner: Simulates 2nd and 3rd Hr After Charge Door is Closed.

Weight %	P. Cake	Plates	MMdust	MMDr	Coke	Fe Chip	Air	Oxygen	Lenox	NiterDs	Silica	Anzon	NatGas
PbO	42.9%	19.0%	35.3%	0.0%	0.0%	0.0%	0.00%	0.00%	40.1%	10.0%	0.00%	71.1%	0.0%
Pb*	8.0%	73.2%	0.0%	87.3%	0.0%	0.0%	0.00%	0.00%	0.0%	61.2%	0.0%	0.0%	0.00%
C*	12.00%	1.50%	0.00%	2.3%	92.2%	3.00%	0.00%	0.00%	0.0%	0.50%	0.00%	22.1%	75.00%
CO ₂	6.00%	0.00%	0.00%	0.00%	0.00%	0.00%	0.00%	0.00%	0.0%	0.00%	0.00%	0.00%	0.00%
Fe*	0.00%	0.1%	0.00%	0.00%	0.00%	82.50%	0.00%	0.00%	0.0%	0.00%	0.00%	0.0%	0.00%
FeO	2.80%	0.1%	0.50%	0.50%	0.29%	12.00%	0.00%	0.00%	0.9%	0.50%	0.30%	0.0%	0.00%
SO ₃	1.00%	2.0%	12.0%	0.00%	0.00%	0.00%	0.00%	0.00%	0.0%	0.00%	0.00%	4.0%	0.00%
S	0.00%	0.00%	0.00%	1.70%	5.50%	0.00%	0.00%	0.00%	0.3%	1.50%	0.00%	0.00%	0.00%
Na(K)2O	1.10%	0.00%	35.0%	2.00%	0.10%	0.00%	0.00%	0.00%	3.5%	7.30%	0.01%	0.00%	0.00%
SiO ₂	11.00%	1.00%	0.50%	1.50%	1.50%	2.50%	0.00%	0.00%	36.0%	0.50%	99.6%	0.99%	0.00%
Ca(Ba)O	1.50%	0.15%	0.10%	0.60%	0.09%	0.00%	0.00%	0.00%	0.90%	5.00%	0.00%	1.50%	0.00%
MgO	7.50%	0.00%	0.10%	0.02%	0.04%	0.00%	0.00%	0.00%	0.70%	0.10%	0.00%	0.00%	0.00%
Cr2O ₃	1.10%	0.00%	0.13%	0.02%	0.00%	0.00%	0.00%	0.00%	0.00%	0.00%	0.00%	0.00%	0.00%
Al2O ₃	5.00%	0.05%	0.10%	0.05%	0.18%	0.00%	0.00%	0.00%	16.1%	0.00%	0.10%	0.00%	0.00%
NaCl	0.06%	0.52%	10.0%	1.00%	0.10%	0.00%	0.00%	0.00%	0.26%	0.50%	0.00%	0.35%	0.00%
SnO	0.01%	0.10%	2.50%	0.60%	0.00%	0.00%	0.00%	0.00%	0.10%	6.00%	0.00%	0.00%	0.00%
Sb2O ₃	0.02%	2.30%	1.50%	1.20%	0.00%	0.00%	0.00%	0.00%	0.10%	6.50%	0.00%	0.00%	0.00%
Other	0.00%	0.00%	0.00%	0.00%	0.00%	0.00%	0.00%	0.00%	0.00%	0.00%	0.00%	0.00%	0.00%
ZnO	0.02%	0.00%	2.00%	0.20%	0.00%	0.00%	0.00%	0.00%	1.00%	0.20%	0.00%	0.00%	0.00%
SO ₂	0.000%	0.000%	0.000%	0.000%	0.000%	0.000%	0.00%	0.00%	0.000%	0.000%	0.000%	0.000%	0.000%
O ₂	0.000%	0.000%	0.000%	0.000%	0.000%	0.000%	23.20%	100.0%	0.000%	0.000%	0.000%	0.000%	0.000%
CuO	0.01%	0.00%	0.25%	1.00%	0.00%	0.00%	0.00%	0.00%	0.00%	0.20%	0.00%	0.00%	0.00%
H ₂ OXL	0.00%	0.00%	0.00%	0.00%	0.00%	0.00%	0.00%	0.00%	0.00%	0.00%	0.00%	0.00%	0.00%
CO	0.000%	0.000%	0.000%	0.000%	0.000%	0.00%	0.00%	0.00%	0.000%	0.000%	0.000%	0.000%	0.000%
H ₂	0.000%	0.000%	0.000%	0.000%	0.000%	0.00%	0.00%	0.00%	0.000%	0.000%	0.000%	0.000%	25.0%
N ₂	0.000%	0.000%	0.000%	0.000%	0.000%	76.8%	0.0%	0.000%	0.000%	0.000%	0.000%	0.000%	0.00%
Sum:	100.0%	100.0%	100.0%	100.0%	100.0%	100.0%	100.0%	100.0%	100.0%	100.0%	100.0%	100.0%	100.0%
H₂O Fr.	0.0%	0.0%	0.0%	0.0%	0.0%	0.0%	0.5%	0.0%	0.0%	0.0%	0.0%	0.0%	0.0%
	P. Cake	Plates	MMdust	MMDr	Coke	Fe Chip	Air	Oxygen	Lenox	NiterDs	Silica	Anzon	NatGas
OrigH ₂ O	35.0%	2.0%	1.0%	3.0%	3.0%	1.8%	0.5%	0.0%	23.0%	3.0%	2.0%	0.0%	0.0%

Moles	P. Cake	Plates	MHDust	MMDrs	Coke	Fe Chip	Air	Oxygen	Lenox	NiterDs	Silica	Anzor	NatGas	
PbO	0.0023	0.0051	0.0007	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	
Pb ⁺	0.0005	0.0214	0.0000	0.0241	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	
C ⁻	0.0118	0.0076	0.0000	0.0110	0.0236	0.0006	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0067	
CO ₂	0.0016	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	
Fe ⁺	0.0000	0.0001	0.0000	0.0000	0.0000	0.0033	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	
FeO	0.0005	0.0001	0.0000	0.0004	0.0000	0.0004	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	
SO ₃	0.0001	0.0015	0.0007	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	
S	0.0000	0.0000	0.0000	0.0030	0.0005	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	
Na(K)2O	0.0002	0.0000	0.0025	0.0018	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	
SiO ₂	0.0022	0.0010	0.0000	0.0014	0.0001	0.0001	0.0000	0.0000	0.0000	0.0000	0.0000	0.0147	0.0000	
Ca(Ba)O	0.0003	0.0002	0.0000	0.0006	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	
MgO	0.0022	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	
Cr2O ₃	0.0001	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	
Al ₂ O ₃	0.0006	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	
NaCl	0.0000	0.0005	0.0008	0.0010	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	
SnO	0.0000	0.0000	0.0001	0.0003	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	
Sb ₂ O ₃	0.0000	0.0005	0.0000	0.0002	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	
H ₂ OFrCb	0.0000	0.0000	0.0000	0.0000	0.0000	0.0003	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	
ZnO	0.0000	0.0000	0.0001	0.0001	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	
SO ₂	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	
O ₂	0.0000	0.0000	0.0000	0.0000	0.0000	0.0068	0.0107	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	
CuO	0.0000	0.0000	0.0000	0.0007	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	
H ₂ OXL	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	
CO	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	
H ₂	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0134	
N ₂	0.0000	0.0000	0.0000	0.0000	0.0000	0.0257	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	
Sum:	H ₂ O Fr.	P. Cake	Plates	MHDust	MMDrs	Coke	Fe Chip	Air	Oxygen	Lenox	NiterDs	Silica	Anzor	NatGas

Master Metal's 3.6-m SRF Using 100% O₂ to Burner: Simulates 2nd and 3rd Hr After Charge Door is Closed.

ESTIMATE OF STACK EMISSIONS (FROM NEW BAGHOUSE) WHEN OPERATING THE 3.6-M ROTARY FURNACE,

and Summary of Results of Furnace and Flue Calculations

	Drying	Melting	Reduction	
Operation time during each furnace charging cycle:	Hr	2.0	Hr	
Initial temperature for drying, melting, or reduction:	°F	590	°F	
Temperature at finish of drying, melting, or reduction:	°F	1445	°F	
Natural gas firing rate:	Btu/Hr	5.6 MM	Btu/Hr	
Average off-gas temperature from the furnace:	°F	1796	°F	
Natural gas consumed by the furnace:	Ft ³ /Hr	5245	Ft ³ /Hr	
Oxygen consumed by the furnace:	Ft ³ /Hr	8953	Ft ³ /Hr	
Diluted flue gas to the stack				
Flow rate entering stack @ 95 °F :	ACFM	12,888	ACFM	
Flow, DSCFM = 77°F @ 1 atm.:	DSCFM	11,746	DSCFM	
Moisture content of gas to stack:	Vol%	1.9%	Vol%	
Particulates to stack, grains/dscf:	g/DSCF	0.0026	g/DSCF	
SO ₂ to the stack:	Lb/Hr	0.12	Lb/Hr	
Emissions from Stack (Lb/Hr)				
Tetraethyl lead:	*BDL	Lb/Hr	*BDL	Lb/Hr
Total chlorine expressed as chloride:		Lb/Hr	0.016	Lb/Hr
Lead as Pb:		Lb/Hr	0.085	Lb/Hr
Antimony as Sb:		Lb/Hr	0.0032	Lb/Hr
Arsenic as As:	**ID	Lb/Hr	**ID	Lb/Hr
Barium as Ba:		Lb/Hr	<0.01	Lb/Hr
Silver as Ag:		Lb/Hr	<0.01	Lb/Hr
Cadmium as Cd:		Lb/Hr	<0.01	Lb/Hr
Chromium as Cr		Lb/Hr	0.0002	Lb/Hr
Mercury, thallium, and beryllium:	***NIF		***NIF	***NIF
* Below Detectable Limit	** Insufficient data to estimate.		*** Not known to be in feedstock to plant.	

Master Metal's 3.6-m SRF Using 100% O2 to Burner: Simulates 2nd and 3rd Hr After Charge Door is Closed.

mt	Bullion	Mat+Slag	Dust Out	mt	All mt In	All mt Out	MoleChecks	At. Wt.		
PbO	0.00000	1.81183	0.15851	PbO	1.81183	1.97034	0.054066	223		PbO
Pb*	9.36265	0.00000	0.00000	Pb*	9.50978	9.36265	0.054066	207		Pb*
C*	0.00000	0.61204	0.00000	C*	0.73447	0.61204	0.933348	12		C*
CO2	0.00000	0.00000	0.65849	CO2	0.72921	1.17814	0.933348	44		CO2
Fe*	0.00000	0.00000	0.00000	Fe*	0.18665	0.00000	0.00474	55.8		Fe*
FeO	0.00000	0.33805	0.00224	FeO	0.10013	0.34030	0.00474	71.8		FeO
S03	0.00000	0.00000	0.05378	S03	0.18648	0.05378	0.005897	80		S03
S	0.00000	0.16715	0.00000	S	0.11410	0.16715	0.005897	32		S
Na20	0.00000	1.05556	0.15707	Na(K)20	1.21263	1.21263	RefrConsum	62		Na(K)20
Si02	0.00000	1.17083	0.00224	Si02	1.17308	1.17308	0.00000	60		Si02
Ca0	0.00000	0.06132	0.00045	Ca(Ba)0	0.06176	0.06176	0.00000	56		Ca(Ba)0
Mg0	0.00000	0.08966	0.00045	Mg0	0.09011	0.09011	0.00000	40.3		Mg0
Cr203	0.00000	0.01411	0.00058	Cr203	0.01469	0.01469	0.00000	152		Cr203
Al203	0.00000	0.06625	0.00045	Al203	0.06670	0.06670	0.00000	102		Al203
NaCl	0.00000	0.09744	0.04488	NaCl	0.14232	0.14232	0.00000	58.5		NaCl
Sn0	0.00000	0.04046	0.01122	Sn0	0.05168	0.05168	Sulfide	135		Sn0
Sb203	0.00000	0.20779	0.00673	Sb203	0.21452	0.21452	339.0	291		Sb203
H20FrCb	0.00000	0.00000	0.00000	H20FrCb	0.00469	0.24611	Atomic	18		H20FrCb
Zn0	0.00000	0.01167	0.00898	Zn0	0.02064	0.02064	97.4	81.4		Zn0
S02	0.00000	0.00000	0.00006	S02	0.00000	0.00006	Weights	64		S02
O2	0.00000	0.00000	0.00001	O2	0.56081	0.03440	86.9	32		O2
Cu0	0.00000	0.05728	0.00112	Cu0	0.05840	0.05840	159.0	79.5		Cu0
H20XL	0.00000	0.00000	0.00000	H20XL	0.00000	0.00000		18		H20XL
CO	0.00000	0.00000	0.00000	CO	0.00000	0.00000		28.00		CO
H2	0.00000	0.00000	0.00000	H2	0.02682	0.00000		2.00		H2
N2	0.00000	0.00000	0.00000	N2	0.71984	0.71984		28.00		N2
Sum:	9.3626	5.8014	1.107267	Sum:	17.79136	17.79135	CHECK			
	Bullion	Mat+Slag	Dust Out							
			1.107267							
			Check							
							02Mole Σ			
							1.35951			
							1.359499			

Master Metal's 3.6-m SRF Using 100% O₂ to Burner: Simulates 2nd and 3rd Hr After Charge Door is Closed.

Moles To Flux	
0.0074	PbO
0.0459	Pb ⁺
0.0612	C ⁻
0.0166	CO ₂
0.0033	Fe ⁺
0.0014	FeO
0.0017	SO ₃
0.0036	S
0.0170	Na(K)2O
0.0195	SiO ₂
0.0011	Ca(Ba)O
0.0022	MgO
0.0001	Cr ₂ O ₃
0.0006	Al ₂ O ₃
0.0017	NaCl
0.0003	SnO
0.0007	Sb ₂ O ₃
0.0137	H ₂ OFrCb
0.0001	ZnO
0.0000	SO ₂
0.0175	O ₂
0.0007	CuO
0.0000	H ₂ OXL
0.0000	CO
0.0134	H ₂
0.0257	N ₂
Moles To Flux	

Master Metal's 3.6-m SRF Using 100% O₂ to Burner: Simulates 2nd and 3rd Hr After Charge Door is Closed.

	mt	Total In	Total Out	mt	At. Wt.	m3/tmole	
PbO	1.8118	1.9703		PbMoleSum	223	25.93	PbO
Pb*	9.5098	9.3626		11.1916	207	32.34	Pb*
C*	0.7345	0.6120		11.1916	12	5.33	C*
CO ₂	0.7292	1.1781		Fe&S m.s.	44	0.00	CO ₂
Fe*	0.1866	0.0000		0.00474	55.8	14.68	Fe*
FeO	0.1001	0.3403		0.00474	71.8	15.28	FeO
S0 ₃	0.1865	0.0538		0.00590	80		S0 ₃
S	0.1141	0.1672		0.00590	32	16.00	S
Na(K)2O	1.2126	1.2126		Refractory	62	31.00	Na(K)2O
SiO ₂	1.1731	1.1731		0.00000	60	27.27	SiO ₂
Ca(Ba)O	0.0618	0.0618		0.00000	56	18.67	Ca(Ba)O
MgO	0.0901	0.0901		0.00000	40.3	12.21	MgO
Cr2O ₃	0.0147	0.0147		0.00000	152	34.55	Cr2O ₃
Al2O ₃	0.0667	0.0667		0.00000	102	31.88	Al2O ₃
NaCl	0.1423	0.1423		0.00000	58.5	34.41	NaCl
SnO	0.0517	0.0517		C moleSum	135	20.93	SnO
Sb2O ₃	0.2145	0.2145		0.07778	291	55.96	Sb2O ₃
H2OFrCb	0.0047	0.2461		0.07778	18	0.00	H2OFrCb
ZnO	0.0206	0.0206		Hwt Sum	81.4	14.88	ZnO
S0 ₂	0.0000	0.0001		0.02735	64		S0 ₂
O ₂	0.5608	0.0344		0.02735	32		O ₂
CuO	0.0584	0.0584		0 wt Sum	79.5	12.42	CuO
H2OXL	0.0000	0.0000		1.35950	18		H2OXL
CO	0.0000	0.0000		1.35951	28.0		CO
H2	0.0268	0.0000		Check?	2.0		H2
N2	0.7198	0.7198			28.0		N2
Sum:	17.7914	17.7914			At. Wt.	m3/tmole	
		mt In	mt Out				
							g/cc-ft ³ /lb 62.42

Master Metal's 3.6-m SRF Using 100% O₂ to Burner: Simulates 2nd and 3rd Hr After Charge Door is Closed.

Boundary Conditions for Operating the Short Rotary Furnace		H fusion kcal/kgmole	Fusion Kcal/hr
Dust recycle rate, dry mt/hr	0.45	PbO	6570
Calculated Natl. Gas Kcal/hr Input	1,423,331	Pb°	1150
Calculated Natl. Gas used, mt/hr	0.107	C	
Calculated Natl. Gas used, kg/hr	107.30	CO ₂	
Natl. Gas heat content (net), kcal/kg	13,265	FeS	7730
Natl. Gas density, lb/ft ³ @ STP	0.044	FeO	7500
Fuel heat capacity, kgcal/kg°C	0.55	SO ₃	
Natl. Gas density, kg/m ³ @ STP	1.44	Na(K)2O	11400
Fuel temperature, C	7	SiO ₂	2600
Air temperatures, C	7	Ca(Ba)O	12200
Solids feed conc. temperature, C	7	MgO	18500
Return dust temperature, C	7	Cr ₂ O ₃	4000
Offgas temperature, C	785	Al ₂ O ₃	25700
Final solids temperature, °C	785	NaCl	6700
Stoic. oxygen for fuel, kg dry/kg fuel	4.00	SnO	6000
% excess oxygen for natural gas	-0.20	Sb ₂ O ₃	13300
Air inleakage to furnace, mt/hr	0.94	CuO	13580
Atomizing steam, kcal/kg	0.00	ZnO	4500
Atomizing steam, kg/kg fuel	0.00	Sum =	141430.
Sum H ₂ O in furnace solids, mt/hr	0.00		356,326
Moisture in air&O ₂ &Gas, mt /hr	0.00	Chemical Reactions	
Total moisture to plant, mt/hr	0.00		
Furnace shell temperature, F	425		
Furnace (small) shell area, m ²	57.65		
Heat loss, kJ/in shell & door, kcal/hr	96,507		
		Large =	6°C

Master Metal's 3.6-m SRF Using 100% O₂ to Burner: Simulates 2nd and 3rd Hr After Charge Door is Closed.

DrossProdmt 0.00233	Dross, %	Normalized Silicate Fraction i.e., Slag Assay	PbO;PbS Pb* C* CO ₂ Fe*;FeS FeO SO ₃ S;Na ₂ S Na(K)2O SiO ₂ Ca(Ba)O MgO Cr2O ₃ Al2O ₃ NaCl SnO Sb2O ₃ H ₂ OFrCb ZnO SO ₂ O ₂ CuO H ₂ OXL CO H ₂ N ₂ SUM	
	0.00000	8.000%		
	0.00000	86.118%		
	0.00000	0.000%		
	0.00000	0.000%		
	0.00000	0.000%	3.1% Al ₂ O ₃ +Cr ₂ O ₃	
	0.00000	0.000%	10.9% FeO	
	0.00000	0.000%	-	
	0.00000	5.000%	-	
	0.00000	0.000%	41.1% Na ₂ O	
	0.00000	0.000%	45.0% SiO ₂	
	0.00000	0.000%	100.0% TOTAL	
	Sulfur Balance (mt or %)			
	0.00000	0.000%	Heat S In as S: 0.1887	Sulfur Pool= 0.04404
	0.00000	0.000%	Heat S Out as S 0.1887	Oxygen Pool= -6.29128
	0.00000	0.140%	Heat SO ₂ as S: 0.000189	FeSvsNa ₂ Smat 50.0%
	0.00000	0.583%	SDross(Heat) 0.0000	Na ₂ O pool= 0.63538
	0.00000	0.000%	Ssilicate(Heat) 0.1138	
	0.00000	0.000%	Smatte(Heat) 0.0534	
	0.00000	0.000%	S in dust=Heat 0.0216	
	0.00000	0.000%	S out by addn. 0.1889	
	0.00000	0.159%		
	0.00000	0.000%		
	0.00000	0.000%	FlueGasMt/hr 1.5201	
	0.00000	0.000%		
	0.00000	0.000%	Wt% SO ₂ 0.0000%	
	0.00000	100.0%	SO ₂ exit/hr, Short Tons this furnace: 0.000062	01
		SO ₂ exit/yr, st/yr all furnaces: 0.42		
		% of S as Silicate 60.3%		
		% of S as Matte 28.3%		
		% of S as Dross		
		% of S as Dust 11.5%	Total% = Check	
		% of S as Gas 0.10%	100%	

Master Metal's 3.6-m SRF Using 100% O₂ to Burner: Simulates 2nd and 3rd Hr After Charge Door is Closed.

mt of CO ₂ =	0.658	
O from PbO=	0.000	but not dust.
O for Fe=	0.000	but not dust.
O from SO ₃ =	0.000	but not dust.
O from SnO=	0.0000	but not dust.
O from Sb ₂ O ₃ =	0.0000	but not dust.
O from ZnO=	0.0000	but not dust.
O from CuO=	0.0000	but not dust.
O for Si in Fe ^a	0.0000	
mt eq CO ₂ in=	0.6585	
		FeOMoleFract.
		0.07
	C02/C0wt.=R	1000.00
	1.27R + 1 =	1271.0000

$$(CO_2-x)/1.27x=CO_2/C_0 \text{ wt=R}$$

$$(CO_2-x)=1.27Rx$$

$$CO_2 \text{ decomposed} = 1.27Rx + x = x(1.27R+1)$$

x= 0.001 mt CO₂ decomposed

1.27x= 0.001 mt CO evolved from melt.

CO₂-x= 0.658 mt CO₂ evolved from melt.

0.000 mt Carbon needed for above.

0.370 mt C

0.612 mt carbon for slag; (inc. H₂)

0.34

mt

20 % Excess Coke

Master Metal's 3.6-m SRF Using 100% O₂ to Burner: Simulates 2nd and 3rd Hr After Charge Door is Closed.

Density 11.30 Pb°	mt O ₂ Out	m ³ Matte/Slag	mt O ₂ In	PbO	Pb°	C°	C0 ₂	m ³ of Bullion: 0.83
	PbO	0.14137	0.21068	0.13000	Pb°	Pb°	C°	m ³ of Bullion: 0.83
	Pb°	0.00000	0.00000	0.00000				m ³ of Slag: 1.88
	C°	0.00000	0.27202	0.00000				m ³ of Dross: 0.00
	C0 ₂	0.85683	0.00000	0.53033				m ³ in Charge: 2.71
	Fe°	0.00000	0.00000	0.00000				
	FeO	0.07583	0.07193	0.02231				
	SO ₃	0.03227	0.00000	0.11189				
	S	0.00000	0.08358	0.00000				
	Na(K)2O	0.31294	0.52778	0.31294				
	SiO ₂	0.62564	0.53220	0.62564				
	Ca(Ba)O	0.01765	0.02044	0.01765				
	MgO	0.03578	0.02717	0.03578				
	Cr203	0.00464	0.00321	0.00464				
	Al203	0.03139	0.02070	0.03139				
	NaCl	0.00000	0.05732	0.00000				
	SnO	0.00612	0.00627	0.00612				
	Sb203	0.03539	0.03996	0.03539				
	H2OFrCb	0.21460	0.00000	0.00000				
	ZnO	0.00406	0.00213	0.00406				
	SO ₂	0.00003	0.00000	0.00000				
	O ₂	0.03441	0.00000	0.56081				
	CuO	0.01175	0.00895	0.01175				
	H2OXL	0.00000	0.00000	0.00000				
	CO	0.00000	0.00000	0.00000				
	H ₂	0.00000	0.00000	0.00000				
	N ₂	0.00000	0.00000	0.00000				
	Sum = Oxygen Out =	2.44069	1.88433	2.44069	<- Oxygen In			
			SUM OF M ₃	Oxygen In				

Master Metal's 3.6-m SRF Using 100% O₂ to Burner: Simulates 2nd and 3rd Hr After Charge Door is Closed.

Heat Capacity Calc., kcal/kg			Mass Basis mt/hr	Heat Inputs kcal/hr		Mass Basis mt/hr	Heat Outputs kcal/hr	
T1, C=	310	for solids						
T2, C=	785	for solids						
Average Feed	0.180	4.3235E-05	Fuel Combustion	0.107	1,423,331	Sensible Heats:		
C0 ₂	253	kcal/kg	Steam for Atomization	0.00	0	C0 ₂ exhaust dry gas	131,363	
CO	257		Coke Combustion:	0.0131	100,980	CO exhaust dry gas	0	
Dust recy.	0.184	0.0000838	Sensible Heats:			O2 exhaust dry gas	8,039	
T1, C=	25.	for gases	Dry furnace feedstocks	7.97	408,909	N2 exhaust dry gas	184,035	
T2, C=	980.	for gases	Dry dust recycle	0.22	11,767	S0 ₂ exhaust dry gas	10	
O2	234	kcal/kg	Natural gas	0.107	416	Solids in Charge (heated/hr)	1,289,479	
S0 ₂	179		Gas combustion oxygen	0.34	562	Dry dust in offgas	34,739	
Air	0.234	0.0000173	Total air into furnace	0.94	1,535	Slag discharge is zero	0	
H2O vapor	489	kcal/kg	H2O in all input gases	0.005	33	H2O from all input gases	2,289	
N2	256		Free H2O in solid feeds	0.00	0	Free H2O in solid feeds	0	
H2O Vaporize	598	Heat of boiling	FeS-Fe°&FeO Formation ΔH	0		0.00		
			Na2Sfrom PbS	"	0	H2O from combustion	117,941	
			Na2Sfrom S0 ₃	"	0	0.24		
			Na2O-Na2C0 ₃	"	0	H2O from atomization steam	0	
			Pb° from PbO	"	0	0.00		
			Refractory dissolved:	0.0000	0	H2O from solid Rx product:	0	
			Heats of fusion and vaporization:			0.00		
			Fusion	-55,134				
			Vaporization=already in ΔH	0		Shell+Door+Refr heat losses	124,504	
						Total Heat Out, kcal/hr:	1,892,399	

Master Metal's 3.6-m SRF Using 100% O₂ to Burner: Simulates 2nd and 3rd Hr After Charge Door is Closed.

	Slag Wt,mt	Matte Wt,mt		
PbO	1.81183	0.00000	-	10.7% % fixed C in matte + slag.
-	0.00000	0.00000	PbS	0.0% is PbS in Matte
C	0.61204	0.00000	C	is Matte + Slag factor kg Pb loss/mt bullion.
CO ₂	0.00000	0.00000	CO ₂	0.00 is (matte/slag) concentration ratio for coke
-	0.00000	0.06041	FeS	5.0% is the % Na ₂ S in slag
FeO	0.28261	0.00604	FeO	2.03% is the percent sulfur in slag.
	0.00000	0.00000	-	
Na ₂ S	0.27734	0.05367	Na ₂ S	
Na(K)2O	0.79245	0.00000	-	
SiO ₂	1.17066	0.00018	SiO ₂	0.005 is (matte/slag) concentration ratio for silica
Ca(Ba)O	0.06131	0.00001	Ca(Ba)O	0.005 is (matte/slag) concentration ratio for lime
MgO	0.08965	0.00001	MgO	0.005 is (matte/slag) concentration ratio for magnesia
Cr ₂ O ₃	0.01411	0.00000	Cr ₂ O ₃	0.005 is (matte/slag) concentration ratio for chromia
Al ₂ O ₃	0.06624	0.00001	Al ₂ O ₃	0.005 is (matte/slag) concentration ratio for alumina
NaCl	0.09695	0.00050	NaCl	0.17 is (matte/slag) concentration ratio for chloride
SnO	0.03986	0.00060	CdO	0.50 is (matte/slag) concentration ratio for tin
Sb ₂ O ₃	0.20612	0.00195	Sb ₂ O ₃	0.27 is (matte/slag) concentration ratio for antimony
H ₂ OFrCb	0.00000	0.00000	H ₂ OFrCb	
ZnO	0.01133	0.00041	ZnS	1.00 is (matte/slag) concentration ratio for Zn
SO ₂	0.00000	0.00000	SO ₂	
O ₂	0.00000	0.00000	O ₂	
CuO	0.01430	0.04298	Cu ₂ S	100.00 is (matte/slag) concentration ratio for Cu
-	0.00000	0.00000	-	
-	0.00000	0.00000	-	
-	0.00000	0.00000	-	
-	0.00000	0.00000	-	
TOTAL SLAG	5.54678	0.16677	TOTAL Matte	
	Slag Wt,mt	Matte Wt,mt		

Master Metal's 3.6-m SRF Using 100% O₂ to Burner: Simulates 2nd and 3rd Hr After Charge Door is Closed.

Calculation of Slag Viscosity			Wt %	Weight	MoleFrac.Slag	MoleNonMatte
X' SiO ₂ ' =	Moles Si+P+Ti	PbO	38.0%	1.81	0.1479	0.00812
X' CaO' =	Moles Ca+Mg+Mn+Pb+Zn+Fe	Pb*	0.0%	0.00	0.0000	0.00000
X' Al ₂ O ₃ ' =	Moles Al ₂ O ₃ +Cr ₂ O ₃	C*	0.0%	0.00	0.0000	0.00000
X' Na ₂ O' =	Moles Na ₂ O+Na ₂ S (non-mat)	CO ₂	0.0%	0.00	0.0000	0.00000
X' NaF ₂ ' =	Moles NaCl	Fe*	0.0%	0.00	0.0000	0.00000
A' =		FeO	5.9%	0.28	0.0716	0.00394
B' =		SO ₃	0.0%	0.00	0.0000	0.00000
		Na ₂ S	2.4%	0.11	0.0647	0.00356
		Na(K)20	16.6%	0.79	0.2326	0.01278
		SiO ₂	24.5%	1.17	0.3551	0.01951
		Ca(Ba)O	1.3%	0.06	0.0199	0.00109
		MgO	1.9%	0.09	0.0405	0.00222
		Cr ₂ O ₃	0.3%	0.01	0.0017	0.00009
		Al ₂ O ₃	1.4%	0.07	0.0118	0.00065
		NaCl	2.0%	0.10	0.0302	0.00166
		SnO	0.8%	0.04	0.0054	0.00030
		Sb ₂ O ₃	4.3%	0.21	0.0129	0.00071
		Other	0.0%	0.00	0.0000	0.00000
		ZnO	0.2%	0.01	0.0025	0.00014
		SO ₂	0.0%	0.00	0.0000	0.00000
		O ₂	0.0%	0.00	0.0000	0.00000
		CuO	0.3%	0.01	0.0033	0.00018
		H ₂ OXL	0.0%	0.00	0.0000	0.00000
		CO	0.0%	0.00	0.0000	0.00000
		H ₂	0.0%	0.00	0.0000	0.00000
		N ₂	0.0%	0.00	0.0000	0.00000
		Sum:	100.00%	4.77	1.0000	0.05495

Master Metal's 3.6-m SRF Using 100% O₂ to Burner: Simulates 2nd and 3rd Hr After Charge Door is Closed.

			100	Start-up controller		
Calculation of the low-temperature heat capacities						
	A	T ₂ - T ₁	B	T _{2sq} - T _{1sq}	C	1/T ₂ -1/T ₁
CO ₂	10.57	25	1.05E-03	1.43E+04	2.06E+05	-3.07E-04
CO	6.708	25	5.53E-04	1.43E+04	6.20E+03	-3.07E-04
Sb406 vap.	52.016	25	1.69E-03	1.43E+04	8.30E+05	-3.07E-04
O ₂	7.16	25	5.00E-04	1.43E+04	4.00E+04	-3.07E-04
SO ₂	11.04	25	9.40E-04	1.43E+04	1.84E+05	-3.07E-04
H ₂ O vapor	7.3	25	1.23E-03	1.43E+04	0.00E+00	-3.07E-04
N ₂	6.83	25	4.50E-04	1.43E+04	1.20E+04	-3.07E-04

Master Metal's 3.6-m SRF Using 100% O₂ to Burner: Simulates 2nd and 3rd Hr After Charge Door is Closed.

Calculation of high-temperature heat capacities, kcal/kg							Low-Temp. Heat Capacities:	
	A	T ₂ - T ₁	B	T ₂ sq - T ₁ sq	C	1/T ₂ -1/T ₁	T ₁ , °C = 0	T ₂ , °C = 25
CO ₂	10.57	955	1.05E-03	1.48E+06	2.06E+05	-2.6E-03	CO ₂	4.91E+00
CO	6.708	955	5.53E-04	1.48E+06	6.20E+03	-2.6E-03	CO	6.20E+00
Σ mass for input heat calc:								
Σ mass for output heat calc								
Σ mass in from material bal:								
Sb406 vapor	52.016	955	1.69E-03	1.48E+06	8.30E+05	-2.6E-03	Sb203 vapor	1.83E+00
O ₂	7.16	955	5.00E-04	1.48E+06	4.00E+04	-2.6E-03	O ₂	5.43E+00
SO ₂	11.04	955	9.40E-04	1.48E+06	1.84E+05	-2.6E-03	SO ₂	3.64E+00
H ₂ O vapor	7.3	955	1.23E-03	1.48E+06	0.00E+00	-2.6E-03	H ₂ O vapor	1.11E+01
N ₂	6.83	955	4.50E-04	1.48E+06	1.20E+04	-2.56E-03	N ₂	6.20E+00

Master Metal's 3.6-m SRF Using 100% O₂ to Burner: Simulate Hrs 4 to 6 After Charge Door is Closed (Complete the Reduction)

Furnace:	Old (3.6-m)	Pb° in Bulln, mt	11.55	CHARGE CYCLE	Bullion	Slag Only	Matte Only
Date:	23-Jul-91	Pb in Feed, mt	11.76	IS FIRST	PbO	2.0%	
Final Charge T°	925 degC	Pb in slag, mt	0.065	*****	Pb*, PbS	100.0%	7.0%
Parameters to Check		Pb in Dust, mt	0.15	TIME IS DURING	C*	4.1%	8.2%
Slag FeO =	9.5%	Pb,Dross,Float,mt	0.00	HRS 4-5-6	CO2	0.0%	0.0%
Target =	10 to 15%	Bullion, mt	11.55	3	FeS		33.3%
Basicity =	1.13	PbToBulln&Dross	98.2%	TO	FeO	7.3%	3.3%
Target =	1.1 to 1.4	Balanced if small	-5.80E-08	COMPLETE	S		
Viscosity =	30	Silicate Phase, mt	3.44	REDUCTION	Na2S	5.0%	29.6%
Target =	10 to 40	Matte Phase, mt	0.36		Na(K)2O	24.6%	
Slag Ratio of		Soda/hr to flue	0.045		SiO2	36.8%	0.2%
(Si+Al)/Na =	1.4	Dust, % of feed	2.9%	*****	Ca(Ba)O	2.1%	0.01%
Molds Output-->		Molds Output-->	8.2	6.3	MgO	4.0%	0.02%
Feed(Scoops)	Dry Feed, lb	Dry Feed Mt/hr		Assay of Slag, %	Cr2O3	0.6%	0.00%
P.CakeTEL(2)	4000.	1.81		Pb	2.3%		
MM Plate(6)	13600.	6.17		C*	4.5%	A1203	2.9%
Dust In (2)	1000.	0.45		FeO	9.5%	NaCl	2.9%
MM Dross(6)	13000.	5.89		S	4.5%	Sn*, SnO	0.6%
Coke(1/2)	700.	0.32		Na2O	28.1%	0.0%	0.0%
FeChips(0.35)	500.	0.23		SiO2	33.3%	Sb*, Sb2S3	0.0%
Total Air In	1842.	0.83		Ca(Ba)O	1.9%	Other	0.0%
O2 to Burner	751.	0.34		MgO	3.6%	ZnO,ZnS	0.3%
Lenox(0)	0.	0.00		Cr2O3	0.6%	Other	0.0%
NiterDross(0)	0.	0.00		Al2O3	2.6%	Other	0.0%
Glass Sand(2)	2000.	0.91		NaCl	2.6%	Cu,Cu2S	0.0%
Anzon(0)	0.	0.0		SnO	1.0%	TOTAL	100.0%
Natural Gas	234.7	0.106		Sb2O3	4.2%	MgCr Brick, kg/hr per	99.90%
TOTAL	37628.	17.05		Other	0.0%	mt Bullion	
Flux Summary							
Soda(2+BH)	3500.	1.586		ZnO	0.3%	0.0	Coke in matte
Sand (1.5)	2000.	0.907		Other	0.0%		+ slag, %
FeChip(.25)	500.	0.227		CuO	1.6%	0.3	"Slag" Basicity
Coke (0.5)	700.	0.317					1.13
Comments: Reduction is completed over a 2-hr period. The decomposition of soda ash is very endothermic.							
Viscosity of the silicate slag, Poise: 30							
DrossRecycle: 0.00 SO2 controller							

Master Metal's 3.6-m SRF Using 100% O₂ to Burner: Simulate Hrs 4 to 6 After Charge Door is Closed (Complete the Reduction)

Gas firing rate:	Summary of Heat Losses, Kcal/hr During 3rd part of 1st charge cycle			Approx. st SO ₂ vented from plant per yr	Estimated ΔT drop due to heat loss when gas passes through the knockout tanks prior to initial dilutn.
Kcal/hr	1,411,292	Loss through furnace walls	127,638	0.43	167 degC
Btu/hr	5,600,008	Heat to refractory from system	21,617	SCFM to Stack:	16064.
Kg Nat. Gas/hr	106.4	Radiant heat loss thru closed door	7,704		
Lb NG/hr	234.7	Sum of heat losses:	156,960		
% TongeO ₂ Use	100.%	Initial T°C of refractory (Ave.)	735 degC	Gas Temp. exiting stack:	35 degC
% Exc.O ₂ ->NG	-20%	Final T°C of refractory (Ave.)	750 degC	Gas Temp. exiting convective cooler:	345 degC
Ave. Offgas T°C	1180 degC	Coke Content of Oxide Component:	4.5%	Gas Temp. exiting convective cooler:	195 degC
P, inch H ₂ O	0.25	Furnace area in use, m ²	57.7	ACFM to BH:	5,995
Shell, T°F	500 degF	Temperature of bag house:	105 degC	SCFM to BH:	4,539
Tramp Air in Through the Door, kg/hr:	835.	PbInSilicatePhaseOfEmulsion, mt	0.070	Matte/slag concentration ratio for coke:	2.00
Percent of Dust Exiting Per Hour	16.7%	Pb in Matte phase of Emulsion, %:	7.0%	SCFM moisture to stack:	242
Percent of Charge Fused Per Hour	12.8	Hrs of 1st Cycle	Furnace Offgas Volume % Undiluted		
		CycleDoorClosed	02:	3.5%	Number of Hrs in this period:
			N2:	38.7%	Soda ash CO ₂ still in solids?
		Total Hrs Burner	H2O:	23.0%	Soda ash CO ₂ to gas vent?
		15.0	CO2:	34.9%	Coke heat input, Kcal/hr
Stoic. Pure O ₂ for Natural Gas, kg/hr	425.6	% Dross Formed?	CO:	0.0%	System's heat gain from refractory, Kcal/hr
Stoic. Pure O ₂ for solid C-> CO ₂ , Kg/hr	361.4	0	SO ₂ :	0.000346%	Heat lost to furnace offgas & dust, Kcal/hr
			sum=	100.00%	Heat to furnace charge from burner, Kcal/hr
					Heat lost from walls and door, Kcal/hr
Initial Temperature of Charge:	785 degC				Sum of heats of reaction and fusion:
		ff x 100°C, +0			Air drawn into furnace via door, SCFM
Final Temperature of Charge:	925 degC	0			382
		ff x 100°C, +1			FurnaceGas/kg " ",kgMole
Charge Moisture Removed:	0.0%		02:	0.07	1853. 59.0
Offgas volume at average furnace temperature:			N2:	0.64	Gas Volume to Baghouse, ACMH = 10,186
	7,317 ACMH		H2O:	0.24	
Dilution air to cooler, kg/hr	3,360		CO2:	0.91	v/oCO ₂ diluted 1.84%
Dilutn. air to cooler, ACMH	6,152		CO:	0.0000	From Stack
Total ACMH into cooler	13,469		SO ₂ :	0.0000	v/oSO ₂ diluted 0.00002%
			sum=	1.8528	
					EFFICIENCY OF HEAT UTILIZATION, %
Comments:	Note that the exit gas temperature is the average during the period. Apparent offgas temperature is low, due to large endothermic reactions			Space between door and furnace, inches =	0.39
				Start-up Controller for totals =	100
				SO ₂ , %/hr	3.3%
				Lb SO ₂ /Hr	SO ₂ evoln.,hrs
				0.029	3.0
				Press., mm:	730
				Ambient, degC:	7

Master Metal's 3.6-m SRF Using 100% O₂ to Burner: Simulate Hrs 4 to 6 After Charge Door is Closed (Complete the Reduction)

(reduction & Na ₂ CO ₃ decomposition) depressing rate of charge temperature increase.					Efficiency
Kg moles of gas from furnace	59.0	Oxide Slag mt	6.399	Kcal to solids	275,034
Kg moles air cooler entry	116.5	345 degC	C with Oxide	Kcal to redn. & fusion	439,718
Kg moles gas into cooler	175.5	195 degC	%C in Oxide	Kcal to H ₂ O evaporation	0
Air to cooler exit, kg/hr	4038.	105 degC	C Burnt, mt	Kcal from coke excess	-125,666
Air to cooler exit, ACMH	4521.		C in dust, mt		
Kg moles air to cooler exit	140.0	Dilution air to stack, kg/hr:	22745.	TOTAL	589,086
Kg moles, all gases, to BH	315.5	Total gas to stack, kg/Moles/hr	1116.	Burner Kcal	1,411,292
ACMH to BH, TOTAL	10186.	Total ACFM into stack	17285.	Efficiency	41.7%
	Atomic Weight	Total Moles (inc. refractory)	Total mt	O ₂ released by redn., mt:	0.1461
				SCFH air drawn into furnace:	22,924
				SCFM air drawn into furnace:	382
PbO	223	0.00946	2.109		
Pb ^a	207	0.04737	9.805		
C*, non-gas	12	0.06208	0.745		
CO ₂	44	0.00247	0.109		
Fe*	55.8	0.00341	0.190		
FeO	71.8	0.00166	0.120		
SO ₃	80	0.00245	0.196		
S	32	0.00368	0.118		
Na(K)2O	62	0.00479	0.297		
SiO ₂	60	0.02111	1.267		
Ca(Ba)O	56	0.00130	0.073		
MgO	40.3	0.00342	0.138		
Cr ₂ O ₃	152	0.00014	0.022		
Al ₂ O ₃	102	0.00099	0.101		
NaCl	58.5	0.00235	0.137		
SnO	135	0.00039	0.053		
Sb ₂ O ₃	291	0.00075	0.220		
H ₂ OFrCb	18	0.00023	0.004		
ZnO	81.4	0.00026	0.021		
SO ₂	64.0	0.00000	0.000		
O ₂	32.0	0.01666	0.533		
CuO	79.5	0.00076	0.060		
Sum:		0.18573	16.316		

Master Metal's 3.6-m SRF Using 100% O₂ to Burner: Simulate Hrs 4 to 6 After Charge Door is Closed (Complete the Reduction)

Weight %	P. Cake	Plates	MMdust	MMDrs	Coke	Fe Chip	Air	Oxygen	Lenox	NiterDs	Silica	Anzon	NatGas
PbO	42.9%	19.0%	35.3%	0.0%	0.0%	0.0%	0.00%	0.00%	40.1%	10.0%	0.00%	71.1%	0.0%
Pb ⁺	8.0%	73.2%	0.0%	87.3%	0.0%	0.0%	0.00%	0.00%	0.0%	61.2%	0.0%	0.0%	0.00%
C ⁺	12.00%	1.50%	0.00%	2.3%	92.2%	3.00%	0.00%	0.00%	0.0%	0.50%	0.00%	22.1%	75.00%
CO ₂	6.00%	0.00%	0.00%	0.00%	0.00%	0.00%	0.00%	0.00%	0.0%	0.00%	0.00%	0.00%	0.00%
Fe ⁺	0.00%	0.1%	0.00%	0.00%	0.00%	82.50%	0.00%	0.00%	0.0%	0.00%	0.00%	0.0%	0.00%
FeO	2.80%	0.1%	0.50%	0.50%	0.29%	12.00%	0.00%	0.00%	0.9%	0.50%	0.30%	0.0%	0.00%
SO ₃	1.00%	2.0%	12.0%	0.00%	0.00%	0.00%	0.00%	0.00%	0.0%	0.00%	0.00%	4.0%	0.00%
S	0.00%	0.00%	0.00%	1.70%	5.50%	0.00%	0.00%	0.00%	0.3%	1.50%	0.00%	0.00%	0.00%
Na(K)2O	1.10%	0.00%	35.0%	2.00%	0.10%	0.00%	0.00%	0.00%	3.5%	7.30%	0.01%	0.00%	0.00%
SiO ₂	11.00%	1.00%	0.50%	1.50%	1.50%	2.50%	0.00%	0.00%	36.0%	0.50%	99.6%	0.99%	0.00%
Ca(Ba)O	1.50%	0.15%	0.10%	0.60%	0.09%	0.00%	0.00%	0.00%	0.90%	5.00%	0.00%	1.50%	0.00%
MgO	7.50%	0.00%	0.10%	0.02%	0.04%	0.00%	0.00%	0.00%	0.70%	0.10%	0.00%	0.00%	0.00%
Cr ₂ O ₃	1.10%	0.00%	0.13%	0.02%	0.00%	0.00%	0.00%	0.00%	0.00%	0.00%	0.00%	0.00%	0.00%
Al ₂ O ₃	5.00%	0.05%	0.10%	0.05%	0.18%	0.00%	0.00%	0.00%	16.1%	0.00%	0.10%	0.00%	0.00%
NaCl	0.06%	0.52%	10.0%	1.00%	0.10%	0.00%	0.00%	0.00%	0.26%	0.50%	0.00%	0.35%	0.00%
SnO	0.01%	0.10%	2.50%	0.60%	0.00%	0.00%	0.00%	0.00%	0.10%	6.00%	0.00%	0.00%	0.00%
Sb ₂ O ₃	0.02%	2.30%	1.50%	1.20%	0.00%	0.00%	0.00%	0.00%	0.10%	6.50%	0.00%	0.00%	0.00%
Other	0.00%	0.00%	0.00%	0.00%	0.00%	0.00%	0.00%	0.00%	0.00%	0.00%	0.00%	0.00%	0.00%
ZnO	0.02%	0.00%	2.00%	0.20%	0.00%	0.00%	0.00%	0.00%	1.00%	0.20%	0.00%	0.00%	0.00%
SO ₂	0.000%	0.000%	0.000%	0.000%	0.000%	0.000%	0.00%	0.00%	0.000%	0.000%	0.000%	0.000%	0.000%
O ₂	0.000%	0.000%	0.000%	0.000%	0.000%	0.000%	23.20%	100.0%	0.000%	0.000%	0.000%	0.000%	0.000%
CuO	0.01%	0.00%	0.25%	1.00%	0.00%	0.00%	0.00%	0.00%	0.00%	0.20%	0.00%	0.00%	0.00%
H ₂ OXL	0.00%	0.00%	0.00%	0.00%	0.00%	0.00%	0.00%	0.00%	0.00%	0.00%	0.00%	0.00%	0.00%
CO	0.000%	0.000%	0.000%	0.000%	0.000%	0.000%	0.00%	0.00%	0.000%	0.000%	0.000%	0.000%	0.000%
H ₂	0.000%	0.000%	0.000%	0.000%	0.000%	0.000%	0.00%	0.00%	0.000%	0.000%	0.000%	0.000%	25.0%
N ₂	0.000%	0.000%	0.000%	0.000%	0.000%	0.000%	76.8%	0.0%	0.000%	0.000%	0.000%	0.000%	0.00%
Sum:	100.0%	100.0%	100.0%	100.0%	100.0%	100.0%	100.0%	100.0%	100.0%	100.0%	100.0%	100.0%	100.0%
H ₂ O Fr.	0.0%	0.0%	0.0%	0.0%	0.0%	0.0%	0.5%	0.0%	0.0%	0.0%	0.0%	0.0%	0.0%
P. Cake	Plates	MMdust	MMDrs	Coke	Fe Chip	Air	Oxygen	Lenox	NiterDs	Silica	Anzon	NatGas	
OrigH ₂ O	0.0%	0.0%	0.0%	0.0%	0.0%	0.0%	0.5%	0.0%	0.0%	0.0%	0.0%	0.0%	0.0%

Moles	P. Cake	Plates	MMdust	MNdr's	Coke	Fe Chip	Air	Oxygen	Lenox	NiterDs	Silica	Amzon	NatGas	
PbO	0.0035	0.0053	0.0007	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	
Pb"	0.0007	0.0218	0.0000	0.0249	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	
C'	0.0181	0.0077	0.0000	0.0113	0.0244	0.0006	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0066	
CO ₂	0.0025	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	
Fe"	0.0000	0.0001	0.0000	0.0000	0.0000	0.0034	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	
FeO	0.0007	0.0001	0.0000	0.0004	0.0000	0.0004	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	
SO ₃	0.0002	0.0015	0.0007	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	
S	0.0000	0.0000	0.0000	0.0031	0.0005	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	
Na(K)20	0.0003	0.0000	0.0026	0.0019	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	
SiO ₂	0.0033	0.0010	0.0000	0.0015	0.0001	0.0001	0.0000	0.0000	0.0000	0.0000	0.0000	0.0150	0.0000	
Ca(Ba)0	0.0005	0.0002	0.0000	0.0006	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	
MgO	0.0034	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	
Cr2O ₃	0.0001	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	
Al2O ₃	0.0009	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	
NaCl	0.0000	0.0005	0.0008	0.0010	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	
SnO	0.0000	0.0000	0.0001	0.0003	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	
Sb2O ₃	0.0000	0.0005	0.0000	0.0002	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	
H2OFrCb	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0002	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	
ZnO	0.0000	0.0000	0.0001	0.0001	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	
SO ₂	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	
O ₂	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0060	0.0106	0.0000	0.0000	0.0000	0.0000	0.0000	
CuO	0.0000	0.0000	0.0000	0.0000	0.0007	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	
H2OXL	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	
CO	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	
H ₂	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0133	0.0000	
N ₂	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	0.0228	0.0000	0.0000	0.0000	0.0000	0.0000	0.0000	
Sum:	H ₂ O Fr.	P. Cake	Plates	MMdust	MNdr's	Coke	Fe Chip	Air	Oxygen	Lenox	NiterDs	Silica	Amzon	NatGas

ESTIMATE OF STACK EMISSIONS (FROM NEW BAGHOUSE) WHEN OPERATING THE 3.6-M ROTARY FURNACE, and Summary of Results of Furnace and Flue Calculations					
	Drying	Melting	Reduction		
Operation time during each furnace charging cycle:	Hr	Hr	3.0	Hr	
Initial temperature for drying, melting, or reduction:	°F	°F	1445	°F	
Temperature at finish of drying, melting, or reduction:	°F	°F	1697	°F	
Natural gas firing rate:	Btu/Hr	Btu/Hr	5.6 MM	Btu/Hr	
Average off-gas temperature from the furnace:	°F	°F	2156	°F	
Natural gas consumed by the furnace:	Ft ³ /Hr	Ft ³ /Hr	5201	Ft ³ /Hr	
Oxygen consumed by the furnace:	Ft ³ /Hr	Ft ³ /Hr	8877	Ft ³ /Hr	
Diluted flue gas to the stack					
Flow rate entering stack @ 95 °F :	ACFM	ACFM	17,285	ACFM	
Flow, DSCFM = 77°F @ 1 atm.:	DSCFM	DSCFM	15,822	DSCFM	
Moisture content of gas to stack:	Vol%	Vol%	1.5%	Vol%	
Particulates to stack, grains/dscf:	g/DSCF	g/DSCF	0.0025	g/DSCF	
SO ₂ to the stack:	Lb/Hr	Lb/Hr	0.029	Lb/Hr	
Emissions from Stack (Lb/Hr)					
Tetraethyl lead:	*BDL	Lb/Hr	*BDL	Lb/Hr	*BDL Lb/Hr
Total chlorine expressed as chloride:		Lb/Hr		Lb/Hr	0.021 Lb/Hr
Lead as Pb:		Lb/Hr		Lb/Hr	0.11 Lb/Hr
Antimony as Sb:		Lb/Hr		Lb/Hr	0.0043 Lb/Hr
Arsenic as As:	**ID	Lb/Hr	**ID	Lb/Hr	**ID Lb/Hr
Barium as Ba:		Lb/Hr		Lb/Hr	<0.01 Lb/Hr
Silver as Ag:		Lb/Hr		Lb/Hr	<0.01 Lb/Hr
Cadmium as Cd:		Lb/Hr		Lb/Hr	<0.01 Lb/Hr
Chromium as Cr		Lb/Hr		Lb/Hr	0.0003 Lb/Hr
Mercury, thallium, and beryllium:	***NIF		***NIF		***NIF
* Below Detectable Limit		** Insufficient date to estimate.		*** Not known to be in feedstock to plant.	

Master Metal's 3.6-m SRF Using 100% O₂ to Burner: Simulate Hrs 4 to 6 After Charge Door is Closed (Complete the Reduction)

mt	Bullion	Mat+Slag	Dust Out	mt	All mt In	All mt Out	MoleChecks	At. Wt.			PbO
PbO	0.00000	0.07000	0.16011	PbO	2.10899	0.23011	0.056826	223			PbO
Pb*	11.54928	0.00000	0.00000	Pb*	9.80521	11.54928	0.056826	207			Pb*
C*	0.00000	0.17079	0.00000	C*	0.32812	0.17079	0.417651	12			C*
CO2	0.00000	0.00000	0.00000	CO2	0.32829	0.90517	0.417651	44			CO2
Fe*	0.00000	0.00000	0.00000	Fe*	0.19007	0.00000	0.005071	55.8			Fe*
FeO	0.00000	0.36181	0.00227	FeO	0.11951	0.36408	0.005071	71.8			FeO
S03	0.00000	0.00000	0.05438	S03	0.19583	0.05438	0.006124	80			S03
S	0.00000	0.17421	0.00000	S	0.11763	0.17421	0.006124	32			S
Na2O	0.00000	1.06609	0.15866	Na(K)2O	1.22475	1.22475	RefrConsum	62			Na(K)2O
SiO2	0.00000	1.26437	0.00227	SiO2	1.26664	1.26664	0.00154	60			SiO2
CaO	0.00000	0.07209	0.00045	Ca(Ba)O	0.07254	0.07254	0.00000	56			Ca(Ba)O
MgO	0.00000	0.13730	0.00045	MgO	0.13775	0.13775	0.00000	40.3			MgO
Cr203	0.00000	0.02112	0.00059	Cr203	0.02171	0.02171	0.00000	152			Cr203
Al203	0.00000	0.10048	0.00045	Al203	0.10093	0.10093	0.00231	102			Al203
NaCl	0.00000	0.10024	0.04533	NaCl	0.14557	0.14557	0.00000	58.5			NaCl
SnO	0.00000	0.04170	0.01133	SnO	0.05304	0.05304	Sulfide	135			SnO
Sb203	0.00000	0.21288	0.00680	Sb203	0.21968	0.21968	339.0	291			Sb203
H20FrCb	0.00000	0.00000	0.00000	H20FrCb	0.00415	0.24354	Atomic	18			H20FrCb
ZnO	0.00000	0.01215	0.00907	ZnO	0.02121	0.02121	97.4	81.4			ZnO
S02	0.00000	0.00000	0.00001	S02	0.00000	0.00001	Weights	64			S02
O2	0.00000	0.00000	0.00000	O2	0.53320	0.06603	86.9	32			O2
CuO	0.00000	0.05911	0.00113	CuO	0.06024	0.06024	159.0	79.5			CuO
H20XL	0.00000	0.00000	0.00000	H20XL	0.00000	0.00000		18			H20XL
C0	0.00000	0.00000	0.00000	C0	0.00000	0.00000		28.00			C0
H2	0.00000	0.00000	0.00000	H2	0.02660	0.00000		2.00			H2
N2	0.00000	0.00000	0.00000	N2	0.63804	0.63804		28.00			N2
Sum:	11.5493	3.8643	0.453309	Sum:	17.71970	17.71970					
	Bullion	Met+Slag	Dust Out		CHECK						
			0.453309								
			Check								
							O2Mole Σ				
							1.071093				
							1.071093				

Master Metal's 3.6-m SRF Using 100% O₂ to Burner: Simulate Hrs 4 to 6 After Charge Door is Closed (Complete the Reduction)

Moles To Flux	
0.0087	PbO
0.0474	Pb ⁺
0.0687	C ⁺
0.0174	CO ₂
0.0034	Fe ⁺
0.0016	FeO
0.002	SO ₃
0.0037	S
0.0172	Na(K)2O
0.0211	SiO ₂
0.0013	Ca(Ba)O
0.0034	MgO
0.0001	Cr ₂ O ₃
0.0010	Al ₂ O ₃
0.0017	NaCl
0.0003	SnO
0.0007	Sb ₂ O ₃
0.0135	H ₂ OFrCb
0.0001	ZnO
0.0000	SO ₂
0.0167	O ₂
0.0007	CuO
0.0000	H ₂ OXL
0.0000	CO
0.0133	H ₂
0.0228	N ₂
Moles To Flux	

Master Metal's 3.6-m SRF Using 100% O₂ to Burner: Simulate Hrs 4 to 6 After Charge Door is Closed (Complete the Reduction)

	mt	Total In	Total Out	mt	At. Wt.	m3/tmole	
PbO	2.1090	0.2301		PbMoleSum	223	25.93	PbO
Pb*	9.8052	11.5493	11.7629		207	32.34	Pb*
C*	0.3281	0.1708	11.7629		12	5.33	C*
CO2	0.3283	0.9052		Fe&S m.s.	44	0.00	CO2
Fe*	0.1901	0.0000			55.8	14.68	Fe*
FeO	0.1195	0.3641			71.8	15.28	FeO
S03	0.1958	0.0544	0.00612		80		S03
S	0.1176	0.1742	0.00612		32	16.00	S
Na(K)2O	1.2247	1.2247		Refractory	62	31.00	Na(K)2O
SiO2	1.2666	1.2666	0.00154		60	27.27	SiO2
Ca(Ba)O	0.0725	0.0725			56	18.67	Ca(Ba)O
MgO	0.1378	0.1378			40.3	12.21	MgO
Cr2O3	0.0217	0.0217			152	34.55	Cr2O3
Al2O3	0.1009	0.1009	0.00231		102	31.88	Al2O3
NaCl	0.1456	0.1456			58.5	34.41	NaCl
SnO	0.0530	0.0530		C moleSum	135	20.93	SnO
Sb2O3	0.2197	0.2197	0.03480		291	55.96	Sb2O3
H2OFrCb	0.0042	0.2435	0.03480		18	0.00	H2OFrCb
ZnO	0.0212	0.0212		Hwt Sum	81.4	14.88	ZnO
S02	0.0000	0.0000	0.02706		64		S02
O2	0.5332	0.0660	0.02706		32		O2
CuO	0.0602	0.0602		0 wt Sum	79.5	12.42	CuO
H2OXL	0.0000	0.0000	1.07109		18		H2OXL
CO	0.0000	0.0000	1.07109		28.0		CO
H2	0.0266	0.0000		Check?	2.0		H2
N2	0.6380	0.6380			28.0		N2
Sum:	17.7197	17.7197			At. Wt.	m3/tmole	
		mt In	mt Out				
							g/cc-ft3/lb 62.42

Master Metal's 3.6-m SRF Using 100% O₂ to Burner: Simulate Hrs 4 to 6 After Charge Door is Closed (Complete the Reduction)

Boundary Conditions for Operating the Short Rotary Furnace		H fusion kcal/kgmole	Fusion Kcal/hr
Dust recycle rate, dry mt/hr	0.45	PbO	6570
Calculated Natl. Gas Kcal/hr Input	1,411,292	Pb°	1150
Calculated Natl. Gas used, mt/hr	0.106	C	
Calculated Natl. Gas used, kg/hr	106.39	CO ₂	
Natl. Gas heat content (net), kcal/kg	13,265	FeS	7730
Natl. Gas density, lb/ft ³ @ STP	0.044	FeO	7500
Fuel heat capacity, kgcal/kg°C	0.55	SO ₃	
Natl. Gas density, kg/m ³ @ STP	1.44	Na(K)2O	11400
Fuel temperature, C	7	SiO ₂	2600
Air temperatures, C	7	Ca(Ba)O	12200
Solids feed conc. temperature, C	7	MgO	18500
Return dust temperature, C	7	Cr ₂ O ₃	4000
Offgas temperature, C	925	Al ₂ O ₃	25700
Final solids temperature, °C	925	NaCl	6700
Stoic. oxygen for fuel, kg dry/kg fuel	4.00	SnO	6000
% excess oxygen for natural gas	-0.20	Sb ₂ O ₃	13300
Air inleakage to furnace, mt/hr	0.83	CuO	13580
Atomizing steam, kcal/kg	0.00	ZnO	4500
Atomizing steam, kg/kg fuel	0.00	Sum =	141430.
Sum H ₂ O in furnace solids, mt/hr	0.00		408,896
Moisture in air&O ₂ &Gas, mt /hr	0.00	Chemical Reactions	
Total moisture to plant, mt/hr	0.00	kcal/kg	
Furnace shell temperature, F	500	PbO + 1/2O = Pb° + 1/2CO ₂	24.5
Furnace (small) shell area, m ²	57.65	PbS + Na ₂ O = Na ₂ S + 1/2O ₂	139.3
Heat loss, kJ/m shell & door, kcal/hr	127,638	Fe° + PbS = FeS + Pb°	21.3
		PbS + FeO = FeS + PbO	145.7
		PbOSO ₃ + Na ₂ O = Na ₂ S + PbO + CO ₂	2276.9
		Na ₂ CO ₃ = Na ₂ O + CO ₂	725.0
		LARGE =	69.08

Master Metal's 3.6-m SRF Using 100% O₂ to Burner: Simulate Hrs 4 to 6 After Charge Door is Closed (Complete the Reduction)

DrossProdmt	Dross, %	Normalized Silicate Fraction i.e., Slag Assay	
0.00000	8.000%		PbO;PbS
0.00000	86.151%		Pb"
0.00000	0.000%		C*
0.00000	0.000%		CO ₂
0.00000	0.000%	4.6% Al ₂ O ₃ +Cr ₂ O ₃	Fe";FeS
0.00000	0.000%	9.5% FeO	FeO
0.00000	0.000%	-	SO ₃
0.00000	5.000%	-	S;Na ₂ S
0.00000	0.000%	38.3% Na ₂ O	Na(K)2O
0.00000	0.000%	47.6% SiO ₂	SiO ₂
0.00000	0.000%	100.0% TOTAL	Ca(Ba)O
0.00000	0.000%		MgO
0.00000	0.000%		Cr ₂ O ₃
0.00000	0.000%	Heat S In as S: 0.1960	Al ₂ O ₃
0.00000	0.000%	Heat S Out as S 0.1960	NaCl
0.00000	0.135%	Heat SO ₂ as S: 0.000196	SnO
0.00000	0.560%	SDross(Heat) 0.0000	Sb ₂ O ₃
0.00000	0.000%	Ssilicate(Heat) 0.0705	H ₂ OFrCb
0.00000	0.000%	Smatte(Heat) 0.1037	ZnO
0.00000	0.000%	S in dust=Heat 0.0218	SO ₂
0.00000	0.000%	S out by addn. 0.1962	O ₂
0.00000	0.154%		CuO
0.00000	0.000%		H ₂ OXL
0.00000	0.000%	FlueGasMt/hr 1.8528	CO
0.00000	0.000%		H ₂
0.00000	0.000%	Wt.% SO ₂ 0.0000%	N ₂
0.00000	100.0%	SO ₂ exit/hr, Short Tons this furnace: 0.000014	SUM
		SO ₂ exit/yr, st/yr all furnaces: 0.43	0
		% of S as Silicate 36.0%	
		% of S as Matte 52.9%	
		% of S as Dross	
		% of S as Dust 11.1%	Total% = Check
		% of S as Gas 0.10%	100%

Master Metal's 3.6-m SRF Using 100% O₂ to Burner: Simulate Hrs 4 to 6 After Charge Door is Closed (Complete the Reduction)

mt of CO ₂ =	0.219	
O from PbO=	0.135	but not dust.
O for Fe=	-0.050	but not dust.
O from SO ₃ =	0.085	but not dust.
O from SnO=	0.0013	but not dust.
O from Sb2O ₃ =	-0.0224	but not dust.
O from ZnO=	0.0002	but not dust.
O from CuO=	0.0000	but not dust.
O for Si in Fe*=	-0.0030	FeOMoleFract. 0.07
mt eq CO ₂ in=	0.4204	CO ₂ /COwt.=R 1.27R + 1 = 1.1148

$$(CO_2-x)/1.27x=CO_2/CO \text{ wt=R}$$

$$(CO_2-x)=1.27Rx$$

$$CO_2 \text{ decomposed} = 1.27Rx + x = x(1.27R+1)$$

x= 0.377 mt CO₂ decomposed

1.27x= 0.479 mt CO evolved from melt.

CO₂-x= 0.043 mt CO₂ evolved from melt.

0.217 mt Carbon needed for above.

0.452 mt C

0.171 mt carbon for slag; (inc. H₂)

0.027	Wet stoic. Coke
mt	20.8 Excess Coke

Master Metal's 3.6-m SRF Using 100% O₂ to Burner: Simulate Hrs 4 to 6 After Charge Door is Closed (Complete the Reduction)

Density 11.30	PbO	mt O ₂ Out	m ³ Matte/Slag	mt O ₂ In	PbO	
Pb°		0.01651	0.00814	0.15132	Pb°	
	Pb°	0.00000	0.00000	0.00000	C°	
	C°	0.00000	0.07591	0.00000	CO ₂	0.65831 1.02
	CO ₂	0.00000	0.00000	0.23876	Fe°	0.00000 1.58
	Fe°	0.00000	0.00000	0.00000	FeO	0.08113 0.00
	FeO	0.08113	0.07698	0.02663	SO ₃	0.03263 2.60
	SO ₃	0.03263	0.00000	0.11750	S	0.00000
	S	0.00000	0.08710	0.00000	Na(K)2O	0.31606 6.9
	Na(K)2O	0.31606	0.53304	0.31606	SiO ₂	0.67554 8.8
	SiO ₂	0.67554	0.57471	0.67554	Ca(Ba)O	
	Ca(Ba)O	0.02073	0.02403	0.02073	MgO	
	MgO	0.05469	0.04161	0.05469	Cr2O ₃	
	Cr2O ₃	0.00686	0.00480	0.00686	Al2O ₃	
	Al2O ₃	0.04750	0.03140	0.04750	NaCl	
	NaCl	0.00000	0.05896	0.00000	SnO	
	SnO	0.00629	0.00647	0.00629	Sb2O ₃	
	Sb2O ₃	0.03624	0.04094	0.03624	H ₂ OFrCb	
	H ₂ OFrCb	0.21278	0.00000	0.00000	ZnO	
	ZnO	0.00417	0.00222	0.00417	SO ₂	
	SO ₂	0.00001	0.00000	0.00000	O ₂	
	O ₂	0.06603	0.00000	0.53320	CuO	
	CuO	0.01212	0.00924	0.01212	H ₂ OXL	
	H ₂ OXL	0.00000	0.00000	0.00000	CO	
	CO	0.00000	0.00000	0.00000	H ₂	
	H ₂	0.00000	0.00000	0.00000	N ₂	
	N ₂	0.00000	0.00000	0.00000		
Sum =	Oxygen Out =	2.24759	1.57555	2.24759	<- Oxygen In	
			SUM OF M3	Oxygen In		

Master Metal's 3.6-m SRF Using 100% O₂ to Burner: Simulate Hrs 4 to 6 After Charge Door is Closed (Complete the Reduction)

Heat Capacity Calc., kcal/kg			Mass Basis mt/hr	Heat Inputs kcal/hr	Mass Basis mt/hr	Heat Outputs kcal/hr
T1, C=	785	for solids				
T2, C=	925	for solids				
Average Feed	0.180	4.3235E-05	Fuel Combustion	1,411,292	Sensible Heats:	
			0.106		C0 ₂ exhaust dry gas	283,532
CO ₂	313	kcal/kg	0.00	0	0.91	
CO	316		Coke Combustion:	125,666	CO exhaust dry gas	0
Dust recy.	0.184	0.0000838	0.0164		0.00	
			Sensible Heats:		O ₂ exhaust dry gas	18,931
T1, C=	25.	for gases	Dry furnace feedstocks	771,019	0.07	
T2, C=	1180.	for gases	5.64		N ₂ exhaust dry gas	199,768
			Dry dust recycle	21,130	0.64	
			0.15		S0 ₂ exhaust dry gas	3
			Natural gas	413	0.000001	
O ₂	287	kcal/kg	0.106		Solids in Charge	1,067,183
S0 ₂	221		Gas combustion oxygen	558	(heated/hr) 5.42	
Air	0.234	0.0000173	0.34		Dry dust in offgas	17,640
H ₂ O vapor	607	Sensible Heat	Total air into furnace	1,361	0.08	
			0.83		Slag discharge is zero	0
N ₂	313		H ₂ O in all input gases	29	0.00	
H ₂ O Vaporize	598	Heat of boiling	0.004		H ₂ O from all input gases	2,520
			Free H ₂ O in solid feeds	0	0.004	
			0.00		Free H ₂ O in solid feeds	0
			FeS-Fe ⁺ &FeO Formation ΔH	-2,711		
			Na ₂ Sfrom PbS	" -7,323	H ₂ O from combustion	145,213
			Na ₂ Sfrom S0 ₃	" -26,839	0.24	
			Na ₂ O-Na ₂ CO ₃	" -383,372	H ₂ O from atomization steam	0
			Pb ⁺ from PbO	" -3,563	0.00	
			Refractory dissolved:	-8	H ₂ O from solid Rx product:	0
			0.0003		0.00	
			Heats of fusion and vaporization:		Shell+Door+Refr heat losses	156,960
			Fusion	-15,902		
			Vaporization=already in ΔH	0		
			Total Heat Input, kcal/hr	1,891,750	Total Heat Out, kcal/hr:	1,891,750

Master Metal's 3.6-m SRF Using 100% O₂ to Burner: Simulate Hrs 4 to 6 After Charge Door is Closed (Complete the Reduction,

	Slag Wt,mt	Matte Wt,mt		
PbO	0.07000	0.00000	-	4.5% % fixed C in matte + slag.
-	0.00000	0.02514	PbS	7.0% is PbS in Matte
C	0.14126	0.02952	C	is Matte + Slag factor kg Pb loss/mt bullion.
CO ₂	0.00000	0.00000	CO ₂	2.00 is (matte/slag) concentration ratio for coke
-	0.00000	0.11952	FeS	5.0% is the % Na ₂ S in slag
FeO	0.25212	0.01195	FeO	2.03% is the percent sulfur in slag.
	0.00000	0.00000	-	
Na ₂ S	0.17181	0.10618	Na ₂ S	
Na(K)2O	0.84513	0.00000	-	
SiO ₂	1.26371	0.00066	SiO ₂	0.005 is (matte/slag) concentration ratio for silica
Ca(Ba)O	0.07205	0.00004	Ca(Ba)O	0.005 is (matte/slag) concentration ratio for lime
MgO	0.13723	0.00007	MgO	0.005 is (matte/slag) concentration ratio for magnesia
Cr ₂ O ₃	0.02111	0.00001	Cr ₂ O ₃	0.005 is (matte/slag) concentration ratio for chromia
Al ₂ O ₃	0.10043	0.00005	Al ₂ O ₃	0.005 is (matte/slag) concentration ratio for alumina
NaCl	0.09849	0.00175	NaCl	0.17 is (matte/slag) concentration ratio for chloride
SnO	0.03963	0.00207	CdO	0.50 is (matte/slag) concentration ratio for tin
Sb ₂ O ₃	0.20703	0.00681	Sb ₂ S ₃	0.27 is (matte/slag) concentration ratio for antimony
H ₂ OFrCb	0.00000	0.00000	H ₂ OFrCb	
ZnO	0.01100	0.00138	ZnS	1.00 is (matte/slag) concentration ratio for Zn
SO ₂	0.00000	0.00000	SO ₂	
O ₂	0.00000	0.00000	O ₂	
CuO	0.00516	0.05395	Cu ₂ S	100.00 is (matte/slag) concentration ratio for Cu
-	0.00000	0.00000	-	
-	0.00000	0.00000	-	
-	0.00000	0.00000	-	
-	0.00000	0.00000	-	
TOTAL SLAG	3.43616	0.35909	TOTAL Matte	
	Slag Wt,mt	Matte Wt,mt		

Master Metal's 3.6-m SRF Using 100% O₂ to Burner: Simulate Hrs 4 to 6 After Charge Door is Closed (Complete the Reduction)

Calculation of Slag Viscosity

X"SiO2"=	0.43	Moles Si + P + Ti
X"CaO"=	0.18	Moles Ca + Mg + Mn + Pb + Zn + Fe
X"Al2O3"=	0.02	Moles Al2O3 + Cr2O3
X"Na2O"=	0.32	Moles NaK2O + Na2S (non-mat)
X"CaF2"=	0.03	Moles NaCl

A=	1.73E-07
B=	1.43E+04

	V in Poise	°K	°C
Viscosity=	30.4	1198	925
Viscosity=	108.7	1073	800
Viscosity=	38.3	1173	900
Viscosity=	16.0	1273	1000
Viscosity=	7.6	1373	1100

	Wt %	Weight	MoleFrac.Slag	MoleNonMatte
PbO	2.2%	0.07	0.0064	0.00031
Pb*	0.0%	0.00	0.0000	0.00000
C*	0.0%	0.00	0.0000	0.00000
CO2	0.0%	0.00	0.0000	0.00000
Fe*	0.0%	0.00	0.0000	0.00000
FeO	7.9%	0.25	0.0710	0.00351
S03	0.0%	0.00	0.0000	0.00000
Na2S	2.2%	0.07	0.0446	0.00220
Na(K)2O	26.5%	0.85	0.2758	0.01363
SiO2	39.6%	1.26	0.4261	0.02106
Ca(Ba)O	2.3%	0.07	0.0260	0.00129
MgO	4.3%	0.14	0.0689	0.00341
Cr2O3	0.7%	0.02	0.0028	0.00014
Al2O3	3.1%	0.10	0.0199	0.00098
NaCl	3.1%	0.10	0.0341	0.00168
SnO	1.2%	0.04	0.0059	0.00029
Sb2O3	6.5%	0.21	0.0144	0.00071
Other	0.0%	0.00	0.0000	0.00000
ZnO	0.3%	0.01	0.0027	0.00014
S02	0.0%	0.00	0.0000	0.00000
O2	0.0%	0.00	0.0000	0.00000
CuO	0.2%	0.01	0.0013	0.00006
H2OXL	0.0%	0.00	0.0000	0.00000
CO	0.0%	0.00	0.0000	0.00000
H2	0.0%	0.00	0.0000	0.00000
N2	0.0%	0.00	0.0000	0.00000
Sum:	100.00%	3.19	1.0000	0.04942

Master Metal's 3.6-m SRF Using 100% O₂ to Burner: Simulate Hrs 4 to 6 After Charge Door is Closed (Complete the Reduction,

			100	Start-up controller		
Calculation of the low-temperature heat capacities						
	A	T ₂ - T ₁	B	T _{2sq} - T _{1sq}	C	1/T ₂ -1/T ₁
CO ₂	10.57	25	1.05E-03	1.43E+04	2.06E+05	-3.07E-04
CO	6.708	25	5.53E-04	1.43E+04	6.20E+03	-3.07E-04
Sb406 vap.	52.016	25	1.69E-03	1.43E+04	8.30E+05	-3.07E-04
O ₂	7.16	25	5.00E-04	1.43E+04	4.00E+04	-3.07E-04
SO ₂	11.04	25	9.40E-04	1.43E+04	1.84E+05	-3.07E-04
H ₂ O vapor	7.3	25	1.23E-03	1.43E+04	0.00E+00	-3.07E-04
N ₂	6.83	25	4.50E-04	1.43E+04	1.20E+04	-3.07E-04

Master Metal's 3.6-m SRF Using 100% O₂ to Burner: Simulate Hrs 4 to 6 After Charge Door is Closed (Complete the Reduction)

Calculation of high-temperature heat capacities, kcal/kg							Low-Temp. Heat Capacities:	
	A	T ₂ - T ₁	B	T _{2sq} - T _{1sq}	C	1/T ₂ - 1/T ₁	T ₁ , °C =	0
							T ₂ , °C =	25
CO ₂	10.57	1155	1.05E-03	2.02E+06	2.06E+05	-2.7E-03	CO ₂	4.91E+00
CO	6.708	1155	5.53E-04	2.02E+06	6.20E+03	-2.7E-03	CO	6.20E+00
Σ mass for input heat calc:								
Σ mass for output heat calc								
Σ mass in from material bal:								
Sb406 vapor	52.016	1155	1.69E-03	2.02E+06	8.30E+05	-2.7E-03	Sb203 vapor	1.83E+00
O ₂	7.16	1155	5.00E-04	2.02E+06	4.00E+04	-2.7E-03	O ₂	5.43E+00
SO ₂	11.04	1155	9.40E-04	2.02E+06	1.84E+05	-2.7E-03	SO ₂	3.64E+00
H ₂ O vapor	7.3	1155	1.23E-03	2.02E+06	0.00E+00	-2.7E-03	H ₂ O vapor	1.11E+01
N ₂	6.83	1155	4.50E-04	2.02E+06	1.20E+04	-2.67E-03	N ₂	6.20E+00